

***Review of FY2001
Development Work for
Vitrification of Sodium Bearing
Waste***

*C. M. Barnes
D. D. Taylor*

September 2002



*Idaho National Engineering and Environmental Laboratory
Bechtel BWXT Idaho, LLC*

Review of FY2001 Development Work for Vitrification of Sodium Bearing Waste

**Charles M. Barnes
Dean D. Taylor**

September 2002

**Idaho National Engineering and Environmental Laboratory
High Level Waste Program Division
Idaho Falls, Idaho 83415**

**Prepared for the
U.S. Department of Energy
Assistant Secretary for Environmental Management
Under DOE Idaho Operations Office
Contract DE-AC07-99ID13727**

ABSTRACT

Treatment of sodium-bearing waste (SBW) at the Idaho Nuclear Technology and Engineering Center (INTEC) within the Idaho National Engineering and Environmental Laboratory is mandated by the Settlement Agreement between the Department of Energy and the State of Idaho. This report discusses significant findings from vitrification technology development during 2001 and their impacts on the design basis for SBW vitrification.

SUMMARY

Waste currently stored in tanks at the Idaho Nuclear Technology and Engineering Center (INTEC), located at the Idaho National Engineering and Environmental Laboratory (INEEL), must be processed into waste forms suitable for permanent disposal as part of a Settlement Agreement between the DOE Idaho Operations Office (DOE-ID), the State of Idaho, and the Department of the Navy. In late FY 2000, a roadmap was prepared outlining the technology development required in order to treat this waste, called “sodium-bearing waste” (SBW), using three different technologies. At the direction of DOE-ID, development of vitrification, one of these three technologies, was vigorously pursued in FY-2001. Results from these development activities are contained in twenty-two documents and summarized in tabular form in this report.

During FY 2001, a baseline flowsheet for SBW vitrification was defined and a mass balance generated. To manage the assumptions used to generate the process mass balance, the process functional requirements, and all other data that will ultimately be part of the design basis for the SBW vitrification process, a database was created. Much of the knowledge gained in FY 2001 was incorporated into the database and the mass balance, and was published in September, 2001 (Ref. 28). However, results from experiments that were performed late in the year or work that extended into early FY 2002, were not incorporated into the database. Thus a thorough review of the development results from FY 2001 was performed in order to capture in a single source both the progress made in filling data gaps and the impact of these results on the database and hence the basis for the SBW vitrification flowsheet.

Significant results were obtained in the areas of feed characterization, feed pretreatment, glass formulation, melter operation, offgas characterization, offgas treatment and secondary waste disposal. Feed characterization data which suggested that the waste may likely contain a higher sulfate content than previously believed together with development of a flowsheet that recycled much of the sulfur in the offgas resulted in development of a glass formulation that tolerates higher sulfur concentration in the feed, capturing a higher percentage of the feed sulfate in the glass. A maximum waste loading in the glass that forms no salt-layer was found, and the effects of different reductant additives, reductant concentrations, and other process parameters on the formation and growth of a salt layer on the melt were determined.

Simulant formulations were developed for both waste from tank WM-180 and a “worst case” composition. Tests were performed that compared glass compositions from simulant melts to those from actual SBW. Pretreatment studies demonstrated a method for removal of sulfate from the feed. Off-gas treatment studies demonstrated NO_x abatement and mercury removal.

Results from FY 2001 development affected over 50 “design basis elements” (DBEs), the basic structural units of the database. The data obtained confirmed initial assumptions for many of the DBEs. However, results from several experiments indicate the need for additional development and possible alteration of the baseline flowsheet. These include:

- The nominal waste loading for WM-180 was determined to be 20 wt% (equivalent to 0.91 wt% SO₃ in glass; waste loading scales with equivalent SO₃ concentration for other SBW liquids). Sulfate removal from the feed offers a means to significantly increase the waste loading, and thereby reduce the volume of glass produced and associated glass storage and disposal costs. Another alternative that has been discussed is to adjust the feed reductant concentration (as needed) to convert sulfur to a volatile form which will not be recycled to the melter.
- In one small research-scale pilot test, a high percentage of mercury present in melter offgas was captured by the scrub system. This result, along with tests of simulated scrub neutralization that yielded precipitated solids, suggest that alternatives to the baseline process for offgas scrub treatment may need to be evaluated.
- A much higher air addition rate to the NOxidizer may be needed than originally assumed. If this is the case an SCR reactor may offer significant savings over the baseline NOxidizer scheme.
- Some of the melter separation factors determined from test data for both volatile and nonvolatile species were significantly different from the baseline mass balance assumptions. The new data should be incorporated into the database and mass balance updated in order to determine whether the impacts of these new separation factors on offgas treatment requirements are significant.
- The method previously used to measure the redox state in glass gives erroneous results when vanadium is present due to interferences between vanadium and iron. An alternate method was developed and shown to accurately estimate the redox state with vanadium present.

CONTENTS

ABSTRACT	iii
SUMMARY	iv
ACRONYMS	vii
1. INTRODUCTION	1
1.1 Background	1
1.2 Report Overview	1
1.2.1 Summary of FY 2001 Data	1
1.2.2 Impacts of FY 2001 Work on Technical Baseline	2
1.2.3 SBW Vitrification Process Status	2
2. FY 2001 DEVELOPMENT WORK SYNOPSES	4
2.1 Tabular Task Summaries	4
2.2 Impacts of Tasks on DBEs	30
3. IMPACTS OF FY 2001 WORK TO DBES	36
4. STATUS OF SBW VITRIFICATION PROCESS	55
4.1 Feed System	55
4.1.1 Feed Rheology, Feed Reactions and Frit Versus Glass Forming Components.....	55
4.1.2 Frit Formulation and Feed Mixing.....	55
4.1.3 Reductant Type and Form.....	56
4.1.4 Feed Pretreatment	56
4.2 Melter and Canister Filling System.....	56
4.2.1 Glass Waste Loading, Glass Formulation and Control of Salt Layer	57
4.2.2 Melter Scale-Up	57
4.2.3 Melter Operating Parameters	57
4.2.4 Melter Partition Factors & Reactions.....	58
4.3 Off-Gas Treatment System.....	59
4.3.1 Film Cooler	59
4.3.2 Off-gas Quench and Scrub System	59
4.3.3 Off-gas Solids Filtration.....	61
4.3.4 NOx Abatement	61
4.3.5 Acid Gas Removal	61
4.3.6 Mercury Removal	61
5. REFERENCES	63
mass balances from rsm-1, ev-16, rsm-2.....	72
Attachment 1	66
Attachment 2	71
Attachment 3	76

TABLES

Table 2-1: Principal results from FY-2001 development tasks.....	5
Table 2-2: Impacts of FY-2001 development tasks to DBEs.....	31

ACRONYMS AND ABBREVIATIONS

BDAT	Best demonstrated available technology
CCC	Centerline canister cooling
CETL	Clemson Environmental Test Laboratory
CSM	Centimeter-scale melter
CVS	Compositional variation studies
DBE	Design basis element
DF	Distribution factor, i.e., mass rate of a component in feed divided by the mass rate of that component in an effluent, usually the offgas
DL	Detection limit
DOE	Department of Energy
DRE	Destruction removal efficiency
DWPF	Defense Waste Processing Facility
EA	Environmental assessment
EIS	Environmental Impact Statement
EVS	Ejector venturi scrubber
ERDL	Environmental Research & Development Laboratory
FC	Film cooler
FTIR	Fourier Transform Infrared Spectroscopy
FY	Fiscal year
GAC	Granular activated carbon
GC-MS	Gas chromatography mass spectroscopy
GFC	Glass-forming chemicals
HEME	High efficiency mist eliminator
HEPA	High efficiency particulate air filter
HIC	High Integrity Container
HVAC	Heating, ventilation and air conditioning
HWLLE	High Level Liquid Waste Evaporator
INEEL	Idaho National Engineering and Environmental Laboratory
INTEC	Idaho Nuclear Technology and Engineering Center
IX	Ion exchange
JH	Joule-heated
LLW	Low Level Waste

MACT	Maximum achievable control technology
NGLW	Newly generated liquid waste
NM	Not meaningful or not measured
NRC	Nuclear Regulatory Commission
NTS	Nevada Test Site
NWCF	New Waste Calcining Facility
OG	Off-gas
PCT	Product consistency test
PFD	Process flow diagram
PNNL	Pacific Northwest National Laboratory
PSD	Particle size distribution
RR	Redox Ratio ($[\text{Fe}^{+2}]/[\text{Fe}_{\text{tot}}]$)
RSM	Research scale melter
SBW-9	The first "target" glass formulation used in FY01 pilot tests at PNNL and CETL. It was based on a feed consisting of SBW from WM-180 with composition as specified in Ref. 35 and GFCs (frit) as follows:

<u>GFCs used</u>	<u>Equivalent frit composition</u>	
H_3BO_3	B_2O_3	15 wt%
Fe_2O_3	Fe_2O_3	10 wt%
$\text{LiOH}\cdot\text{H}_2\text{O}$	Li_2O	5 wt%
SiO_2	SiO_2	65 wt%
$\text{Ca}(\text{OH})_2$	CaOH	5 wt%

(see Table 4.6 of Ref. 11)

SBW-22	The second "target" glass formulation used in late-FY01 pilot test at PNNL and CETL. It was also based on an SBW feed from WM-180 but with composition as specified in Ref. 25. The principal differences between the SBW compositions assumed for SBW-9 and SBW-22 were higher levels of phosphate and sulfate in the latter formulation, reflecting the use of ICP analyses of P and S instead of IC analyses of PO_4^{-3} and SO_4^{-2} .
--------	--

The GFCs (frit) used for SBW-22 was as follows:

<u>GFCs used</u>	<u>Equivalent frit composition</u>	
H_3BO_3	B_2O_3	6.0 wt%
Fe_2O_3	Fe_2O_3	1.5 wt%
$\text{LiOH}\cdot\text{H}_2\text{O}$	Li_3O	6.1 wt%
SiO_2	SiO_2	68 wt%
$\text{Ca}(\text{OH})_2$	CaO	5.0 wt%
$\text{Mg}(\text{OH})_2$	MgO	1.8 wt%

	NaOH	Na ₂ O	4.3 wt%
	V ₂ O ₅	V ₂ O ₅	4.9 wt%
	ZrO ₂	ZrO ₂	2.4 wt%
	(see Table 4.6 of Ref. 17)		
SCR	Selective catalytic reduction		
SFMRF	Slurry Fed Melt Rate Furnace		
SL	Salt layer		
SNCR	Selective non-catalytic reduction		
SRS	Savannah River Site		
SS	Stainless steel		
SVOC	Semi-volatile organic compounds		
TBDB	Technical baseline database		
TCLP	Toxicity Characteristic Leaching Procedure		
TDS	Total dissolved solids		
TFA	Tanks Focus Area		
THC	Total Hydrocarbons		
UDS	Undissolved solids		
UTS	Universal Treatment Standards		
VE	Value engineering		
VLE	Vapor liquid equilibria		
VOC	Volatile organic compounds		
WIPP	Waste Isolation Pilot Plant		
WL	Waste loading		
XRD	X-Ray diffraction		

Review of FY2001 Development Work for Vitrification of Sodium Bearing Waste

1. INTRODUCTION

1.1 Background

Waste currently stored in tanks at the Idaho Nuclear Technology and Engineering Center (INTEC), located at the Idaho National Engineering and Environmental Laboratory (INEEL) must be processed into waste forms suitable for permanent disposal as part of a Settlement Agreement¹ between the DOE Idaho Operations Office (DOE-ID), the State of Idaho, and the Department of the Navy. To treat the liquid waste, DOE-ID is considering the treatment options described in the draft environmental impact statement (EIS).² In the summer of 2000, DOE-ID requested the Tanks Focus Area (TFA) to convene a review team of national experts to independently assess the technical alternatives bounded by the Draft EIS with regard to SBW treatment. The review team recommended vitrification as the preferred SBW treatment option.³ Subsequent to DOE-ID's concurrence with this recommendation Bechtel BWXT, Idaho was directed to vigorously pursue development of SBW vitrification. This report contains a summary of the progress of development activities during Fiscal Year (FY) 2001 and a discussion of the impacts of this work on the development of a process design basis for SBW vitrification.

In late FY 2000 a roadmap was prepared outlining the technology development required in order to treat SBW.⁴ The roadmap identified processing requirements and gave a technical needs assessment which identified uncertainties in each of the treatment technologies under consideration, development targets, definition of development tasks and deliverables, and a schedule for completing those activities. Development activities were prioritized based on estimated impacts of uncertainties on the viability and cost of the treatment process. The development schedule was tailored to address the highest impact uncertainties first. The roadmap was the first attempt to plan the development required to design, construct, and operate a vitrification facility to process SBW.

This report builds on the portion of the roadmap directed at vitrification. Data have been obtained through FY 2001 development activities that reduce some of the uncertainty present a year ago. However, as is common in technology development, some experiments led to discovery of additional issues that impact the performance and cost of the process. Having progressed along the path defined by the roadmap we have a clearer view of what lies ahead.

1.2 Report Overview

1.2.1 Summary of FY 2001 Data

The principal objective of this report is to document, in one place, significant information and insights gained from SBW vitrification development work done during FY 2001. The individual development tasks which are the foundation of this information are described in detail in Refs. 5 to 27. The focus in preparing the current report was to identify and categorize only the salient results from these tasks. These results are summarized in a tabular format in Table 2-1. This table provides "one-stop shopping" for summaries of the technical reports published by the INEEL Environmental Research & Development Laboratory (ERDL) and their subcontractors during FY 2001 in support of SBW vitrification.

1.2.2 Impacts of FY 2001 Work on Technical Baseline

During FY 2001, a database (the Technical Baseline Database or “TBDB”) was created to manage (a) the assumptions used to generate the process mass balance, (b) the assumed process functional requirements, and (c) all other data that will ultimately be part of the design basis for the SBW vitrification process. These items are collectively referred to as “design basis elements” (DBEs). The DBEs which are the central component of the TBDB encapsulate (in discrete portions) the knowledge gained through development activities. In addition to the DBEs, the database was designed to include the process flow diagram (PFD) and mass balance(s) for the process, an overall process description, and for each DBE, a description of its basis (where it came from and why it is accepted) and a validation plan if its basis is not considered firm. Although incomplete, the validation plans and bases associated with the DBEs provide a perspective of the maturity of the SBW vitrification process. In its entirety the TBDB constitutes the complete technical baseline for the process and is the basis for the report, *INEEL SBW Vitrification Process*.²⁸

A secondary objective in preparing the present report was to identify impacts of FY 2001 development work to the technical baseline. With the TBDB as the vehicle for describing the baseline we have examined each of the significant results listed in Table 2-1 to determine which (if any) DBEs it might address. In Table 2-2 all 22 of the FY 2001 development tasks are listed across the top and those DBEs which are impacted by any of these tasks are listed vertically. [A complete list of the DBEs is provided in Attachment 1 (Table A-1). Comparison of the complete list with Table 2-2 indicates that not all the DBEs in the TBDB were addressed by FY 2001 development work.]

Following Table 2-2 we provide a brief discussion of the DBEs impacted listed in the table. In light of the decision by the Department of Energy (DOE) at the time of preparation of this report to terminate further consideration of vitrification as a candidate treatment process for SBW, the discussion in Section 3 is necessarily brief and serves primarily to indicate the authors' judgements relative to whether an acceptable basis has been established for each DBE discussed, and if not, what remains to be done.

1.2.3 SBW Vitrification Process Status

The primary focus of development is to determine acceptable ranges of process and design variables that will ensure the process design requirements are satisfied. When sufficient test data are obtained to show that all design requirements can be met, then the flowsheet for that design can be considered validated. The process design requirements for SBW vitrification include the following:

- The facility must process INTEC wastes
- All waste forms produced must be disposable
- The treatment facility must be licensed, permitted and comply with DOE Orders
- The project and treatment schedule must meet Settlement Agreement commitments
- The feed system must deliver homogeneous, qualified feeds to the melter
- The melter must produce glass in a safe and economical manner qualified for disposal
- The offgas treatment system must remove contaminants from the melter offgas

- The secondary waste treatment system must produce disposable wastes containing species removed from the offgas that cannot be incorporated into the glass waste product
- Waste storage systems must be capable of storing wastes until they can be processed or transported to disposal sites

Though not obvious from the above list, the process design is also constrained by economic viability. For example, a glass with a waste loading of 10% would likely meet all disposal requirements. However, a 10% waste loading would increase glass storage and disposal costs as well as plant operating costs due to the prohibitively long processing schedule that would result. Thus, while the primary development focus is to ensure satisfaction of design requirements, the economic viability of the process must also be considered. As development data are generated and technical hurdles are overcome it is necessary to review the overall process and the associated economics with an eye to minimizing cost if such can be done without compromising the other design requirements.

With this in mind the final objective in preparing this report was to provide a "top-down" view of the process and major elements therein; namely, the melter feed system, the glass formulation and the melter itself, and the offgas system (including the handling of the intermediate and waste streams generating thereby). The focus in this final discussion is to provide a more "global" perspective on the impacts of the development work completed in FY 2001 to the major process components, and identify possible improvements to the baseline process configuration and process variables to improve its economic viability and performance.

2. FY 2001 DEVELOPMENT WORK SYNOPSES

2.1 Tabular Task Summaries

Experimental data and results from FY 2001 testing and studies are encapsulated in two tables shown in this section. Table 2-1 contains summaries of significant results and data from the experiments, and Table 2-2 identifies which DBE the data relate to. In Table 2-1, the results are grouped by report (reference number provided immediately after test title) and broken down by subject or “keywords”. Each row entry in Table 2-1 is given a unique ID number used for reference purposes throughout the report. Table 2-2 uses the ID numbers to map the information from Table 2-1 to the DBEs.

Table 2-1: Principal results from FY-2001 development tasks.

ID	KEYWORDS	DATA	COMMENTS
<i>CETL Test #1: Ref. 5</i>			
A-0	Test description	<ul style="list-style-type: none"> • Melter cross-section: 2088 cm² • Feed rate: 0.265-0.375 L/min (half the rate extrapolated from RSM#1 test) • Glass rate: 350 kg/m²-day = 3.0 lbm/hr-ft² • Heated plenum: 600 C and 350 C temperatures in plenum & exit • 30 wt% waste loading • S concentration in feed 113% of nominal for SBW-9 feed (126% reported on page 35 is an error) • 160 g/L sugar reductant concentration • H₂, O₂, NO, CO, CO₂, N₂O, THC, NO₂, HNO₃, N₂O₄, SO₂, Cl, F, I measured in OG • Frit composition (by wt): B₂O₃(15%), Fe₂O₃(10%), Li₂O(5%), SiO₂(65%), CaO(5%)--(SBW-9) • Test terminated due to electrical problems but sufficient data were gathered to obtain steady state material balance 	<p>Objectives:</p> <ul style="list-style-type: none"> • Demonstrate that an acceptable glass can be made at 30 wt% WL • Provide data for melter mass balance • Observe melt rate, cold cap behavior, feed batching and handling, melter operation, material performance
A-1	Cold cap	<ul style="list-style-type: none"> • No conventional cold cap during test. "Cold cap" consisted of a relatively high proportion of boiling liquid (possibly due to the low oxide content of the feed). Cold cap would melt/boil away in ~5 minutes throughout test after feed cutoff. • Dry cold cap material not present after appearance of salt layer 	<ul style="list-style-type: none"> • Feed rate control based on boiling liquid coverage of melt surface
A-2	Salt layer	<ul style="list-style-type: none"> • Salt phase present after steady conditions established • Significant accumulation (525 gm salt recovered at the end of the run) • Salt phase >90% Na + K + SO₄ (normalized), remainder primarily Mo from corrosion of electrodes 	<ul style="list-style-type: none"> • Steady conditions after 3 melter turnovers and establishment of salt phase on melt surface • Graphite, sugar, polystyrene added to melt to remove salt, but effectiveness questionable • Ground anthracite added to melt appeared to be reactive with salt present
A-3	Glass viscosity	<ul style="list-style-type: none"> • 2.68-3.57 Pa-s @ 1150°C 	<ul style="list-style-type: none"> • Two samples drawn after reaching steady state
A-4	Melter DF (overall)	<ul style="list-style-type: none"> • Range: 10-28 	<ul style="list-style-type: none"> • From PM catch in Table 5.18, using glass rate of 3.16 kg/hr (546 kg/173 hr from page iv)
A-5	Melter DFs (semivolatiles)	<ul style="list-style-type: none"> • Range: 1.5-16.2 • Cl(NM), F(NM) 	<ul style="list-style-type: none"> • From data in Tables 6.1, 6.3 • DF=(feed mass)/(feed mass not in glass)

ID	KEYWORDS	DATA	COMMENTS
		<ul style="list-style-type: none"> Element-specific: B(3.8-16.2), Ba(NM), Ca(4.9-6.3), Cd(NM), Cr(4.5-9.5), Cs(1.5-2), K(2.9-3.4), Li(4.9-9), Na(4.4-10), Pb(NM), Ru(1.7-25), Sr (NM), Zn(11.9-18) 	<ul style="list-style-type: none"> For additional information see Attachment 2
A-6	Melter DFs (volatiles)	<ul style="list-style-type: none"> Range: 2-5.1 Element-specific: P(4.5-5.7), S(2.0-2.1) 	"
A-7	Melter DFs (nonvolatiles)	<ul style="list-style-type: none"> Range: 3.3-8.5 Element-specific: Fe(6.0-8.5), Si(4.8-5.1), Al(5.2-5.7), Ce(3.3-3.6) 	"
A-8	Melter DFs (others)	<ul style="list-style-type: none"> Element-specific: Cu(NM), Mn(NM), Mo(NM), Mg(2.7-23.3), Gd(7.1-11) 	"
A-9	Scrubber DFs	<ul style="list-style-type: none"> Range: 1-2 	<ul style="list-style-type: none"> From data in Tables 6.1, 6.2 (Except Zn) DF= (feed mass not in glass)/(feed mass not in glass or scrub solids or scrub liquid)
A-10	Chloride, Fluoride	<ul style="list-style-type: none"> 98.5%-100+-% of chloride & fluoride measured in glass 	<ul style="list-style-type: none"> Concentrations reported for Cl, F in Tables 5.1, 5.3 believed to be in error
A-11	Glass durability	<ul style="list-style-type: none"> Normalized PCT release rates as % of EA Glass values: B(<8.4%), Li(<15.5%), Na(<13.0%) 	<ul style="list-style-type: none"> Normalization based on analyzed glass compositions (Values <10% desired but not necessary)
A-12	Glass liquidus temperature	<ul style="list-style-type: none"> 921 C 	<ul style="list-style-type: none"> One sample, GLAS-183 (Target value is T_{melt}-100)
A-13	Glass redox	<ul style="list-style-type: none"> Fe⁺²/Fe_{tot} = 0.11-0.18 	<ul style="list-style-type: none"> Range of variation after 1.5 melter turnovers (>0.2 is desirable to limit corrosion and electrode oxidation)
A-14	Melter offgas composition	<ul style="list-style-type: none"> H₂(<0.25%), O₂(11.5%), NO(0.89%), CO(0.28%), CO₂(2.9%), N₂O(0.015%), THC(0.009%), NO₂(0.79%), HNO₃(27 ppm), N₂O₄, N₂O₅ (42 ppm), SO₂(199 ppm), Cl(2.6 ppm), F(0.07 ppm), HI(0.096 ppm), I₂(1.26 ppm), I_{total}(2.6 ppm) Variation in reported data: H₂(16%), O₂(8%), NO(63%), CO(40%), CO₂(99%), N₂O(150%), THC(111%), NO₂(92%), HNO₃(67%), N₂O₄(149%), SO₂(200%), Cl(31%), F(200%), I(85%) 	<ul style="list-style-type: none"> These are "most representative values" from Tables 5.14, 5.15 Concentrations calculated from OG flows and measured concentrations at FC outlet OG data corrected to "wet" conditions using 36.2% H₂O (21.3% in Table 5.15 is error; should be 36.2% at melter outlet based on 21.3% at FC outlet and FC dilution of 1.7) Reported air inleakage ~50% of total OG
A-15	Sulfur partitioning	<ul style="list-style-type: none"> ~57% to glass ~12% to scrub liquor & solids ~28% as gaseous sulfur species ~3% in sulfate salt layer 	<ul style="list-style-type: none"> Partitioning %s are based on the total sulfur actually measured in melter products (Tables 5.11& 6.1), not on sulfur in feed Based on 496-531 ppm SO₂ in wet melter OG Gives ~90% mass balance closure on sulfur
A-16	Corrosion	<ul style="list-style-type: none"> Significant corrosion of molybdenum electrodes occurred 	<ul style="list-style-type: none"> Attributed to under-reduced glass

ID	KEYWORDS	DATA	COMMENTS
A-17	Particle size distribution of offgas aerosol	<ul style="list-style-type: none"> Bimodal distribution Peaks at 0.3-1 μm (49 wt%) and 30-100 μm (17.5 wt%) size ranges 	<ul style="list-style-type: none"> PSD of aerosol exiting FC estimated from visual inspection of SEMs Three particle types: Rods, Angular, Cenospheres
A-18	Solids composition	<ul style="list-style-type: none"> Solid deposits in OG lines primarily alkali oxides and salts [major species: Na (20.6-23.5%), K (6.0-8.9%), B (3.1-4.5%), Li (1.4-1.9%)] UDS in scrub solids mainly Si (3-19%), Ca (7-15.5%), Fe (3.3-6.4%), Al (2.6-4%), Na (1.1-4.1%) 	<ul style="list-style-type: none"> Complete analysis of solid deposits in Table 5.7 Complete analysis of scrub solids in Table 5.9
A-19	Film cooler	<ul style="list-style-type: none"> Film cooler average dilution factor 1.7 Film cooler average outlet temperature $\sim 340^\circ\text{C}$ Film cooler average pressure drop 1.2 in. H_2O 	<ul style="list-style-type: none"> From Table 5.13
A-20	Organics in offgas	<ul style="list-style-type: none"> Major organics in OG include nitrophenol (11 ppmv), unknown C_{11} hydrocarbons (22 ppmv), dodecane (10 ppmv), unknown C_9 hydrocarbon (5 ppmv), unknown siloxane (4 ppmv), octane (4 ppmv) 	<ul style="list-style-type: none"> Others given in Table 5.20
A-21	Melter Heat Balance	<ul style="list-style-type: none"> Melter heat loss (heat removed by jacket cooling water) about 60% of total heat demand; water evaporation 10%, heating of gases 26% 	<ul style="list-style-type: none"> From Table 7.1
A-22	Melter offgas conditions	<ul style="list-style-type: none"> Average plenum temperature 637°C, periods of "low" and "high" plenum temperature also run Average film cooler dilution factor 1.7 Average film cooler outlet temperature 340°C 	<ul style="list-style-type: none"> Gas temperature at film cooler inlet 150°C lower than plenum temperature, the latter being biased upward from radiative heating
A-23	Melter mass balance	<ul style="list-style-type: none"> 83% of feed (excluding NO_3^- and H_2O) accounted for in melter outputs 	<ul style="list-style-type: none"> From Executive Summary
A-24	Nitration of sugar	<ul style="list-style-type: none"> SBW feed simulant (from Clemson-2) with 160 g/L analyzed with FTIR showed no evidence of a C-O-N-O bond, indicating that no nitration of the sugar had occurred since the test (approximately a 4-month period) 	<ul style="list-style-type: none"> Communicated via e-mail from K.J. Perry to D.D. Taylor on 1/23/02
A-25	Slagging, plugging	<ul style="list-style-type: none"> Solids accumulated in the FC and in the OG piping upstream and downstream, to the point of flow obstruction, requiring dismantling and cleaning of the FC and adjacent ducting Solids upstream of FC showed evidence of having been molten and were very tough, difficult to remove Solids downstream of FC were primarily condensed salts and were friable, powder-like, and easier to remove All plugging problems were upstream of first quench. 	<ul style="list-style-type: none"> Recommendations: <ul style="list-style-type: none"> Move melter exit duct from side to top of melter, maximally distant from melt and cold cap surfaces and from the feed stream Make FC "bayonet" style with injection of coolant gas immediately inside FC inlet Eliminate elbows in OG ducting or increase radius of curvature (use "sweep" elbows) Include ports in off gas lines to allow <ul style="list-style-type: none"> (a) cleaning without disassembly, (b) observation of effects of feed rate, plenum temperature, film cooler air flow on buildup rate during testing.
A-26	Foaming	<ul style="list-style-type: none"> No indications of a foaming problem were observed. 	<ul style="list-style-type: none"> Average redox ratio ($\text{Fe}^{+2}/\text{Fe}_{\text{tot}}$) was 0.14

ID	KEYWORDS	DATA	COMMENTS
			<ul style="list-style-type: none"> Other "operating conditions" through which control of foaming was achieved not specified
<i>Mercury Removal Tests with GAC: Ref. 6</i>			
B-0	Test description	<ul style="list-style-type: none"> Test gases: Water+air+HgCl₂ (baseline), OG simulant+HgCl₂ or Hg° - Baseline: H₂O(46%), N₂(45%), O₂(8.9%), HgCl₂(803-943 µg/m³) - Simulated OG: H₂O(46%), N₂(39%), O₂(3.7%), CO₂(3.9%), N₂O(0.14%), NO(990 ppm), CO(100 ppm), SO₂(98 ppm), NO₂ (97 ppm), HCl(21 ppm), HgCl₂(774-886 µg/m³) or Hg°(494 µg/m³) Run time: 14-29 hr Bed residence time: 0.1-0.53 sec (< full-scale residence time) Mass of carbon (Mersorb): 0.2-1.0 g (much smaller than full-scale) >99% of mercury (both elemental and HgCl₂) removed from simulated OG 	<p>Objectives:</p> <ul style="list-style-type: none"> Assess short-term impacts of OG components from SBW vitrification on mercury removal efficiency of sulfur-impregnated carbon (Mersorb)
B-1	Hg Removal Efficiency		<ul style="list-style-type: none"> Range 99.0-99.65%
B-2	Effect of other species on Hg removal	<ul style="list-style-type: none"> Other species in OG (N₂O, NO, NO₂, CO, SO₂, HCl) appear to have no detrimental effect on Hg removal efficiency Hg removal efficiencies for the vitrification OG feed (with above species) ranged from 99.3-99.6% Removal efficiencies for the air-water feed ranged from 99.2-99.3% 	<ul style="list-style-type: none"> Concentrations of "other species" in surrogate higher (for some species by a factor of 100-1000) than expected in treatment plant GAC bed feed
<i>Melt Rate Testing/Reductant Selection: Ref. 7</i>			
C-0	Test description	<ul style="list-style-type: none"> Crucible scale tests at INTEC (500 ml crucible) and Clemson Melt Rate Furnace (8-in diameter crucible) RSM#1, 2 and EV-16 tests discussed briefly Slurry-fed crucible melter (500 ml) was designed, tested, and used to obtain curves of redox ratio vs reductant concentration for SBW-9 simulant Effects of reductant type and temperature on melt behavior studied Four reductants studied: sugar, glycolic acid, corn starch, activated carbon MRF includes controlled lid heat, plenum temperature = 750°C 	<p>Objectives:</p> <ul style="list-style-type: none"> Demonstrate vitrification of SBW with direct slurry feeding without feed preconditioning Determine whether reductant is needed Select a reductant Determine if a sulfate salt layer will develop Determine optimum waste loading and melt rate Develop efficient way to measure redox ratio Investigate electrodes to provide real-time measurement of redox ratio in melter
C-1	Redox	<ul style="list-style-type: none"> Acceptable range for redox ratio is $0.01 < \text{Fe}^{+2}/\text{Fe}_{\text{tot}} < 0.30$ No correlation of redox ratio with temperature or reductant type was noted or reported, apparently due to inadequate control of other redox-affecting variables 	<ul style="list-style-type: none"> Iron acts to buffer redox potential of melt Other multivalent species (Cr, Mn, Ce, Ni, Cu) can upset buffering capacity of Fe

ID	KEYWORDS	DATA	COMMENTS
C-2	Effects of salt layer	<ul style="list-style-type: none"> • Salt can extract phosphate, chromate, molybdate, boron, alkali, alkaline earth, and rare earth metals (including Cs) from glass melt • Promotes Cs volatility • Increases melt rate by enhancing heat transfer from melt to cold cap and withing the cold cap itself • Enhances corrosion of melter components 	<ul style="list-style-type: none"> • From literature review
C-3	Sulfur behavior	<ul style="list-style-type: none"> • Sulfate salt phase observed on top of all crucible samples 	<ul style="list-style-type: none"> • No correlation between mass of sulfate salt and redox state of glass in crucible tests
C-4	Foaming	<ul style="list-style-type: none"> • Foaming most pronounced at lower temperatures (no foam observed above 1000°C) 	<ul style="list-style-type: none"> • Temperature range studied: 700-1150°C
C-5	Reductant type	<ul style="list-style-type: none"> • Corn starch not desirable due to poor dissolution and high viscosity of resulting feed • Glycolic acid not desirable because of low reducing efficiency, inability to impact redox ratio in glass • Activated carbon not desirable because of poor reactivity with gas phase (low reduction of nitrate), poor mixing, and potential for overreduction of glass 	<ul style="list-style-type: none"> • Sugar is preferred reductant
C-6	Redox measurement	<ul style="list-style-type: none"> • Spectrophotometric method for quick, in-lab determination of redox ratio $[\text{Fe}^{+2}/\text{Fe}_{\text{tot}}]$ is described^a • Two probes for real-time redox measurement in glass tested • Second probe gave good results but failed due to corrosivity of melt 	<ul style="list-style-type: none"> • Redox determination method obtained by simplifying method used at PNNL • Need more corrosion-resistant materials to make a practical redox probe or less-corrosive melt
<i>Radioactive Crucible Glass Melt Tests: Ref. 8</i>			
D-0	Test description	<ul style="list-style-type: none"> • Glasses prepared and analyzed using: <ul style="list-style-type: none"> - WM-180 simulant with SBW-9 GFC - WM-180 simulant with SBW-22 GFC - Actual WM-180 waste with SBW-9 GFC - Actual WM-180 waste with SBW-22 GFC • Crucibles not capped during melting • PCT testing performed on rad glasses only • Redox ratio: $\text{Fe}^{+2}/\text{Fe}_{\text{tot}} < 0.04$ for all glasses 	<p>Objectives:</p> <ul style="list-style-type: none"> • Demonstrate remote capabilities for making and analyzing crucible size glass samples with actual SBW • Validate representativeness of WM-180 simulant • Evaluate methods of sampling OG during tests • Characterize durability and quality of glass samples produced
D-1	Representativeness of	<ul style="list-style-type: none"> • Glass compositions using SBW-9 (wt%, cold/rad) Al(4.65/4.65); 	<ul style="list-style-type: none"> • Glass melts contained ~10-20% less mass than

^a See also entry M-2 in table.

ID	KEYWORDS	DATA	COMMENTS
	simulant	<p>B(3.04/3.26); Ca(2.78/2.89); Fe(5.29/5.57); Li(1.52/1.58); K(1.85/1.60); Na(10.9/10.9); Si(20.22/16.68); S(0.41/0.45)</p> <ul style="list-style-type: none"> Glass compositions using SBW-22 (cold/rad, wt %) Al(3.07/2.87); B(1.41/1.51); Ca(2.19/2.31); Fe(1.21/1.09); Li(1.93/2.10); K(1.06/1.05); Na(8.57/10.1); Si(26.33/20.94); S(0.30/0.28) Salt layer observed on both cold and rad SBW-9 glasses; no evidence of salt layer on SBW-22 glasses 	<p>expected from volatile content of feed</p> <ul style="list-style-type: none"> Analysis of SBW feed simulant used with SBW-22 GFC showed significant differences in B(15%), Fe(25%), and K(12%) from reported values for actual SBW from WM-180 Sugar concentration used was slightly lower than final recommendation for SBW-9 Simulant judged be representative of hot waste from comparison of measured wt% of major species (<10% difference for all species except Si (21-26%) and K (2-15%) in one glass sample of each type
D-2	Glass properties	<ul style="list-style-type: none"> PCT leach concentrations of SBW glasses about 1/10 those of EA glasses Normalized PCT releases (g/L): B(1.288-1.715); Li(1.150-1.863); Na(1.399-1.702); Si(0.465-0.846) TCLP, µg/ml <ul style="list-style-type: none"> SBW-9: Cd(0.2); Cr(0.41); Pb(0.75) SBW-22: Cd(0.01); Cr(0.02); Pb(<DL) Radioactive glasses analyzed with XRD and found to be amorphous Sampling scheme was defined but not tested 53-58% of the Cs in the feed was retained by the glass in the SBW-9 and SBW-22 formulations, respectively 	<ul style="list-style-type: none"> Difference in TCLP results between SBW-9 and -22 attributed to testing procedure rather than differences in actual durabilities PCT tests performed on radioactive glass samples only
D-4	Hot gas sampling		
D-5	Cs partitioning		<ul style="list-style-type: none"> Higher Cs retention would be expected in a melter than in these open, small-scale crucible melts
<i>Glass Formulation Development: Ref. 9</i>			
E-0	Test description	<ul style="list-style-type: none"> Two sets of crucible tests performed: (a) dry feed with no reductant, (b) liquid simulant with and without reductant (sugar) Dry Feed Tests <ul style="list-style-type: none"> 100 g samples made using dry batch chemicals to simulate WM-180 SBW plus one of 10 different frit compositions; 25% < WL < 45% Samples heated in crucible to 1150°C, poured on SS plate and air cooled 25 g of each sample heat treated to simulate CCC Analysis using XRD, PCT test, viscosity measurement vs temperature Liquid simulant, Method 1 <ul style="list-style-type: none"> Waste simulant, GFC, sugar, and acid mixed to produce 50 g glass Mixtures dried @100-140°C, transferred to crucible and capped with lids coated with frit, heated to 1150°C, quenched to room temperature Liquid simulant, Method 2 	<p><u>Objectives:</u></p> <ul style="list-style-type: none"> Develop candidate glass composition for scaled melter tests of direct vitrification of WM-180 SBW simulant that satisfies the following constraints: <ul style="list-style-type: none"> Eliminates or minimizes salt layer formation on the glass melt Viscosity: $2 < \eta < 10 \text{ Pa}\cdot\text{s}$ @ 1150°C Liquidus: $T_L \leq 1050^\circ\text{C}$ PCT test: Normalized release (r_i) $< 1 \text{ g/m}^2$ Homogeneity: Homogeneous per test by XRD and optical microscopy @ 100x

ID	KEYWORDS	DATA	COMMENTS
		<ul style="list-style-type: none"> - Similar to Method 1, except (a) dry additives slowly mixed into waste in melting crucible, and (b) crucible lids not used • Frit components included SiO₂, B₂O₃, Li₂O, Fe₂O₃, TiO₂, CaO and BaO • Some tests utilized a variation of Method 1 	
E-1	Glass homogeneity (dry feed)	<ul style="list-style-type: none"> • A salt layer was observed on each melt for frits SBW2 and SBW3 for all waste loadings tested (25-45%) • For frit SBW4, glass was homogeneous at waste loadings of 25% and 30%, at 35% waste loading, salt layer was observed • Salt observed for frits SBW5, SBW6, SBW7 and SBW8 at 35% waste loading, homogeneous glass from SBW-9 showing that by increasing CaO, B₂O₃ and Li₂O at the expense of Fe₂O₃ and SiO₂, a salt layer could be avoided • T_L < 1050°C for all tests (dry feed and liquid feed tests) 	<ul style="list-style-type: none"> • SBW2 and SBW3 frits were formulated to show the effect of Li₂O and B₂O₃ on sulfate solubility
E-2	Liquidus Temperature		
E-3	Viscosity	<ul style="list-style-type: none"> • (Dry Feed Tests) Viscosity measured for SBW2 at 5 WLs (25-45%), for SBW4 at 25% WL; all satisfy constraint except SBW2 at 25% WL • (Liquid Feed Tests) Viscosity constraint met by SBW-9 at 25% and 30% WL, but viscosity at 30% WL is close to limit. Viscosity of SBW4 glass at 25% WL exceeds specification 	<ul style="list-style-type: none"> • See Figures 11, 12 and 22 of report
E-4	PCT Response	<ul style="list-style-type: none"> • (Dry Feed Tests) For 6 glasses (SBW2 at 5 WLs and SBW4 at 25% WL) all met PCT constraint except CCC glass for SBW2 at 45% WL • (Liquid Feed Tests) All SBW4 and SBW-9 samples met PCT constraint 	
E-5	Salt Formation (Liquid Feed Tests)	<ul style="list-style-type: none"> • SBW-9 showed the greatest promise for salt-free glass • Effect of sugar concentration on salt formation is minimal • Effect of nitric acid in feed on salt formation is minimal • Salt formation was generally not seen at waste loadings below 30% 	<ul style="list-style-type: none"> • Sugar varied from 77 to 164 g/liter • Reported effect of sugar concentration on salt formation conflicts with other tests (e.g., see H-2)
E-6	Foaming	<ul style="list-style-type: none"> • As concentration of sugar increased, the intensity of foaming increased during drying but decreased during melting 	<ul style="list-style-type: none"> • 4 tests at different sugar concentrations with SBW4 frit and 35% waste loading, 3 tests with SBW-9 and 30% waste loading and different sugar concentrations, one test with SBW-9 frit and 32% waste loading
<i>Off Gas Treatment Technology Development: Ref. 10</i>			
F-0	Study description	<ul style="list-style-type: none"> • DeNOx methods examined: Staged combustion (NOxidizer), SCR - NOxidizer testing done at MSE - Kinetic modeling of staged combustion, SCR, thermal destruction • Scrubber behavior examined: - VLE behavior of Hg 	<p><u>Objectives:</u></p> <ul style="list-style-type: none"> • Summarize TFA-funded investigations of NOx destruction and OG scrubber system performance

ID	KEYWORDS	DATA	COMMENTS
		<ul style="list-style-type: none"> - Acidic scrubber performance using melter OG containing Hg 	
F-1	Staged Combustion	<ul style="list-style-type: none"> • NOxizer performance: - 92-97% NOx destruction efficiency - 99.7% organics destruction efficiency - Reduction of all Hg in feed to elemental Hg 	<ul style="list-style-type: none"> • Surrogate OG generated by oxidation of nitric acid, vaporization of elemental mercury, injection of chlorine and vaporization of benzene • Additional results in Ref. 14 • Integrated test of SCR with EV-16 melter postponed due to September terrorist attack • Additional kinetic modeling results in Ref. 21
F-2	SCR	<ul style="list-style-type: none"> • SCR system procured and installed at Clemson 	
F-3	Kinetic modeling of NOx destruction	<ul style="list-style-type: none"> • Kinetic modeling results: - Near complete NOx destruction using ammonia in SNCR at 870-1040°C - NOx destruction by heating alone not viable due to low destruction efficiencies and high required temperature 	
F-4	Mercury VLE	<ul style="list-style-type: none"> • Factors found to affect Hg VLE were NO₂ concentration of OG, HNO₃ concentration of liquid, and solution temperature • Cl concentration did not affect Hg VLE 	<ul style="list-style-type: none"> • Additional Hg VLE results in Ref. 16
F-5	Scrub acidity	<ul style="list-style-type: none"> • RSM-2 tests showed decrease in acidity of scrub from 1 to ~0.6 M, increase in the acidity of HEME liquid from 0 to 1 M 	
<i>Virification Demonstration—RSM Test #1: Ref. 11</i>			
G-0	Test description	<ul style="list-style-type: none"> • Melter cross-section: 182 cm² • 30, 32, 35 wt% waste loadings • 135, 149.5, 154.5, 178.1, 197 g/L sugar reductant concentrations • H₂, O₂, N₂, NO, CO, CO₂, N₂O, THC measured in OG • Frit composition (by wt): B₂O₃(15%), Fe₂O₃(10%), Li₂O(5%), SiO₂(65%), CaO(5%)--(SBW-9) 	<p>Objectives:</p> <ul style="list-style-type: none"> • Check for SL in SBW-9 at different waste loadings • Characterize glass redox state and leach resistance at different reductant concentrations • Characterize melter OG
G-1	Melt rate	<ul style="list-style-type: none"> • Melt rate limited by the boil-off rate of water in feed • Glass rate: 7.1-11.1 lbm/hr-ft² • Feed rate: 10.7-16.8 L/hr-ft² • Raw SBW simulant rate: 8.6-13.5 L/hr-ft² 	<ul style="list-style-type: none"> • SBW simulant rate calculated from feed rate using ratio of SBW simulant volume (168 L) to total feed volume (209 L, based on Figure 5.3 in report)
G-2	Energy consumption	<ul style="list-style-type: none"> • Average energy requirement over all runs: 6.1 kW-hr/kg glass 	<ul style="list-style-type: none"> • 2-3x normal requirement for JH melters due to high weight fraction of H₂O in SBW
G-3	Glass durability	<ul style="list-style-type: none"> • Tested Ba, Cr, Pb, Cd. Leachate concentrations below detection for all glasses except highly-reduced one. DLs were 0.2, 1, 4, and 5% of TCLP limits. • [Pb]_{leachate} = 7.4% of TCLP limit at reductant concentration of 197 g/L 	<ul style="list-style-type: none"> • Excessive reductant can degrade leaching performance of glass
G-4	Salt layer	<ul style="list-style-type: none"> • Shorting of melter electrodes occurred at 35% waste loading 	<ul style="list-style-type: none"> • Absence of cold cap and overfeeding a possible cause, allowing buildup of conductive

ID	KEYWORDS	DATA	COMMENTS
			aqueous layer (boiling) on surface of melt <ul style="list-style-type: none"> Corrected by increasing feed sugar concentration--operated 16 hours at 140% nominal feed sulfate without further shorting (however, acceptability of glass redox state under these conditions was not determined)
G-5	Melter DFs (semivolatiles)	<ul style="list-style-type: none"> Range: 3-789 Element-specific: B(20-35), Ba(21-138), Ca(71-120), Cd(11-34), Cr(13-20), Cs (3-5), K(7-9), Li(30-44), Na(19-23), Pb(51), Ru(6-10), Sr(37-54), Zn(82-789) 	<ul style="list-style-type: none"> Film cooler OG was sampled, filtered, scrubbed in acid (H₂SO₄) and base (NaOH). Melter DFs were determined from total filter samples and analyses of impinger solutions. For additional information see Attachment 2
G-6	Melter DFs (volatiles)	<ul style="list-style-type: none"> Range: 4-16 Element-specific: P(12-16), S(4-9) 	"
G-7	Melter DFs (nonvolatiles)	<ul style="list-style-type: none"> Range: 1100-1230 Element-specific: Fe(1100-1230) 	"
G-8	Melter DFs (others)	<ul style="list-style-type: none"> Element-specific: Cu(8-15), Mn(302-402), Mo(89-253) 	"
G-9	Gaseous sulfur	<ul style="list-style-type: none"> Sulfur detected in solids collected on total quartz filter samples Sulfur not detected downstream of total filter in impinger solutions 	<ul style="list-style-type: none"> OG was sampled, filtered, scrubbed in acidic (H₂SO₄) and basic (NaOH) impingers
G-10	Sulfur partitioning	<ul style="list-style-type: none"> 55% to glass 26% collected as soluble sulfates in EVS and HEME solutions 19% unaccounted for 	<ul style="list-style-type: none"> SO₂ was not detected in OG (DL>20 ppm) Unaccounted sulfur implies at most 20 ppm SO₂ in OG; but ~1 ppm is more characteristic of low plenum temperature (<600 C) melters Undetected S in HEME and in EVS solids believed to account for missing sulfur
G-11	Salt layer in melter	<ul style="list-style-type: none"> SL present in melter at all waste loadings (30, 32, 35 wt%) SL composition: Na₂SO₄ (80%), alkali halides (15%), Cr (1%) Increased sugar in feed (197 g/L) mitigated SL growth at elevated S in feed (140% of nominal S at 35 wt% waste loading) 	<ul style="list-style-type: none"> SL found in all test segments, increasing in severity with higher waste loading and sulfur content of feed SL typically 1-3 mm in depth, in non-contiguous pools and primarily in meniscus and near melt protrusions Total mass of SL not quantitated
G-12	Cold cap	<ul style="list-style-type: none"> No evidence of conventional cold cap in these tests 	
G-13	Melter offgas composition	<ul style="list-style-type: none"> 30 wt% WL, 135 g/L sugar in feed: H₂(0.14%), NO(4.60%), CO(1.17%), CO₂(8.20%), N₂O(0.95%), THC(0.09%) 32 wt% WL, 135 g/L sugar in feed: H₂(0.15%), NO(3.70%), CO(1.23%), CO₂(6.86%), N₂O(0.73%), THC(0.09%) 35 wt% WL, 135-178.1 g/L sugar in feed: H₂(0.19-0.43%), 	<ul style="list-style-type: none"> %s at left are based on reported average concentrations in Table 6.2 of report, with He contribution subtracted out Table 6.2 values were back-calculated from measured gas composition at FC exit,

ID	KEYWORDS	DATA	COMMENTS
		NO(2.85-3.17%), CO(1.74-2.26%), CO ₂ (7.13-8.26%), N ₂ O(0.69-0.89%), THC(0.09-0.19%) • CO and H ₂ well below flammability limits	assuming no melter are inleakage • Values for 30, 32 wt% WL are arithmetic means of reported values; ranges for 35 wt% are for the four reported values in table. • OG data in report (Tables 6-1, 6-2) show N ₂ mass balance discrepancies of at least 5-17% for the eight runs listed (assuming zero air in-leakage to melter). Probable error bars on data are thus likely to be at least ±20%.
G-14	Glass redox	• 32 wt% WL, 135 g sugar/L simulant: Fe ⁺² /Fe _{tot} = 0.015-0.016 • 35 wt% WL, 135 g sugar/L simulant: Fe ⁺² /Fe _{tot} = 0.002-0.005 • 35 wt% WL, 154 g sugar/L simulant: Fe ⁺² /Fe _{tot} = 0.023-0.058 • 35 wt% WL, 166 g sugar/L simulant: Fe ⁺² /Fe _{tot} = 0.019-0.029 • 35 wt% WL, 177 g sugar/L simulant: Fe ⁺² /Fe _{tot} = 0.028-0.077 • 35 wt% WL, 184 g sugar/L simulant: Fe ⁺² /Fe _{tot} = 0.127 • 35 wt% WL, 197 g sugar/L simulant: Fe ⁺² /Fe _{tot} = 0.138-0.45	• Redox ratio determined by colorimetric method described in Appendix D of report • Steady conditions only achieved for sugar concentration of 135 g/L simulant
G-15	Volatile species	• 30 wt% WL: NO ₂ (81.3%), NO ₃ (18.5%) B, Ca, Ce, Cu, Fe, Mg, Na, Si, Zn, Zr, F (all <0.2%) • 32 wt% WL: NO ₂ (76.8%), NO ₃ (23.1%) B, Ca, Fe, Mg, Na, Si, Zn, Zr, F (all <0.2%) • 35 wt% WL: NO ₂ (71.1%), NO ₃ (28.8%) B, Ca, Fe, Mg, Na, Si, Zn, F (all <0.2%)	• Impinger solutions were chemically analyzed. Only species listed were detected. • Parenthetical values are %'s of the total mass of dissolved species reported in the impinger solutions
G-16	Glass density	• Glass densities of 2.60, 2.61, and 2.61 gm/cc were reported at 30, 32, and 35 wt% waste loading, respectively	
<i>Sulfur Partitioning During SBW Vitrification: Ref. 12</i>			
H-0	Test description	• Two test series performed: - Segmented (vitrification of slurry feed divided into four test steps; each step examined separately in suitably-designed test) - Integrated (feed vitrification in a single step in one test) • Process parameters varied in integral tests: Reductant type (sugar and glycolic acid), Sugar concentration, Melter gas flow rate, HNO ₃ concentration, Feed rate, S concentration in feed • Test scale: 1 inch diameter melt surface • Nominal feed rate: 1.5 ml/min	Objectives: • Determine how sulfur partitioning is influenced by key process parameters in controlled lab scale system (CSM or "Centimeter-Scale Melter") • Compare with sulfur behavior from pilot scale testing • Determine thermodynamically-favored sulfur species
H-1	Sulfur partitioning during feed drying	• Based on results of dry-out tests, less than 2.5% (and likely none) of the S in the feed is lost to the OG during feed drying	

ID	KEYWORDS	DATA	COMMENTS																																																																																								
		<ul style="list-style-type: none">Increasing the concentration of acid from 1 to 3 molar does not increase the amount of S volatilizedThermodynamic calculations confirm that no S should be volatilized during feed drying																																																																																									
H-2	Sulfur partitioning in melter	<ul style="list-style-type: none">CSM parametric test results (first test is baseline):<ul style="list-style-type: none">Baseline parameters: S_{feed} (feed sulfur conc) = SBW-9 value (RSM#1), Sugar = 135 g/L, Purge (melter gas volumetric flow) = 100 cc/min, Acid = SBW-9 value (RSM#1), Feed = 1.5 ml/min.Effects of process parameters on %S_{ret} (% of S in feed retained in glass), SL (extent of salt layer: 0=>no SL, 4=>severe SL coverage), and RR ($RR=[Fe^{+2}]/[Fe_{\text{tot}}]$) indicated below (<u>box</u> indicates variation from baseline value): <table><thead><tr><th>S_{feed}</th><th>Sugar</th><th>Purge</th><th>Acid</th><th>Feed</th><th>%S_{ret}</th><th>SL</th><th>RR</th></tr></thead><tbody><tr><td>1x</td><td>1x</td><td>1x</td><td>1x</td><td>1x</td><td>73</td><td>0.5</td><td>0.068</td></tr><tr><td>1x</td><td>1x</td><td><u>0.5x</u></td><td>1x</td><td><u>0.5x</u></td><td>73</td><td>1</td><td>0.068</td></tr><tr><td>1x</td><td>1x</td><td><u>10x</u></td><td>1x</td><td>1x</td><td>67</td><td>1.5</td><td>0.068</td></tr><tr><td>1x</td><td>1x</td><td>1x</td><td>1x</td><td><u>0.3x</u></td><td>93</td><td>1</td><td>0.068</td></tr><tr><td>1x</td><td>1x</td><td>1x</td><td>1x</td><td><u>4.8x</u></td><td>71</td><td>4</td><td>0.068</td></tr><tr><td>1x</td><td><u>0x</u></td><td>1x</td><td>1x</td><td><u>0.9x</u></td><td>100</td><td>1</td><td>0.010</td></tr><tr><td>1x</td><td><u>0.5x</u></td><td>1x</td><td>1x</td><td><u>0.9x</u></td><td>96</td><td>1</td><td>0.020</td></tr><tr><td>1x</td><td><u>1.5x</u></td><td>1x</td><td>1x</td><td><u>0.9x</u></td><td>31</td><td>0</td><td>0.800</td></tr><tr><td><u>2x</u></td><td>1x</td><td>1x</td><td>1x</td><td><u>0.9x</u></td><td>46</td><td>2</td><td>0.068</td></tr><tr><td>1x</td><td>1x</td><td>3x</td><td><u>2.4x</u></td><td><u>0.9x</u></td><td>77</td><td>0</td><td>0.068</td></tr></tbody></table> <ul style="list-style-type: none">Thermodynamic calculations show 32.4-99.9% of S in the feed goes to OG as SO₂, 0.04% to OG as SO₃, and 0.01% to glass as Na₂SO₄SO₂ concentration in CSM OG < 2.0 ppm (GC-MS)When SO₂ in melter inlet purge gas was spiked to 90 ppm no change was observed in GC-MS responseEV-16-1999 test: 54% of S in feed partitioned to glassRSM-01-1 test: 50-60% of S in feed partitioned to the glass, ~25% captured by OG system<ul style="list-style-type: none">30% waste loading: 63% S in glass32% waste loading: 56% S in glass35% waste loading: 56% S in glass35% waste loading, 140% S in feed: 51% S in glassEV-16-2000-1 test: 57% of the S in the feed partitioned to the glass, 3%	S_{feed}	Sugar	Purge	Acid	Feed	% S_{ret}	SL	RR	1x	1x	1x	1x	1x	73	0.5	0.068	1x	1x	<u>0.5x</u>	1x	<u>0.5x</u>	73	1	0.068	1x	1x	<u>10x</u>	1x	1x	67	1.5	0.068	1x	1x	1x	1x	<u>0.3x</u>	93	1	0.068	1x	1x	1x	1x	<u>4.8x</u>	71	4	0.068	1x	<u>0x</u>	1x	1x	<u>0.9x</u>	100	1	0.010	1x	<u>0.5x</u>	1x	1x	<u>0.9x</u>	96	1	0.020	1x	<u>1.5x</u>	1x	1x	<u>0.9x</u>	31	0	0.800	<u>2x</u>	1x	1x	1x	<u>0.9x</u>	46	2	0.068	1x	1x	3x	<u>2.4x</u>	<u>0.9x</u>	77	0	0.068	<ul style="list-style-type: none">Thermodynamic calculations do not account for solubility of SO₄²⁻ in glassFor CSM results, where S in glass was determined only by XRF, the %S_{ret} was normalized to the expected ICP value using the reported XRF and ICP values for the second baseline testOnly tests which gave no SL were with 150% nominal sugar concentration and with 240% of nominal acid (under-reduced)Only test where 100% of feed S partitioned to glass was with no sugar at redox ratio of 0.01Redox ratio (RR) estimated from sugar concentration using Figure 4.2 in reportGC-MS is either not responsive to SO₂ in OG or nearly all SO₂ reacts or is consumed prior to exit from melterSulfur partitioning data reported here for EV-16-2000-1 do not agree with that reported in Ref. 5. The latter reports 51% of sulfur in feed and 58% of total sulfur measured in melter effluents as present in the glass (vs 72% reported here). Ref. 5 also reports 3% of the feed sulfur as present in the melter salt layer (vs 7.7% reported here)Reduced feed rate with reduced gas purge rate resulted in increased salt accumulation over baseline.
S_{feed}	Sugar	Purge	Acid	Feed	% S_{ret}	SL	RR																																																																																				
1x	1x	1x	1x	1x	73	0.5	0.068																																																																																				
1x	1x	<u>0.5x</u>	1x	<u>0.5x</u>	73	1	0.068																																																																																				
1x	1x	<u>10x</u>	1x	1x	67	1.5	0.068																																																																																				
1x	1x	1x	1x	<u>0.3x</u>	93	1	0.068																																																																																				
1x	1x	1x	1x	<u>4.8x</u>	71	4	0.068																																																																																				
1x	<u>0x</u>	1x	1x	<u>0.9x</u>	100	1	0.010																																																																																				
1x	<u>0.5x</u>	1x	1x	<u>0.9x</u>	96	1	0.020																																																																																				
1x	<u>1.5x</u>	1x	1x	<u>0.9x</u>	31	0	0.800																																																																																				
<u>2x</u>	1x	1x	1x	<u>0.9x</u>	46	2	0.068																																																																																				
1x	1x	3x	<u>2.4x</u>	<u>0.9x</u>	77	0	0.068																																																																																				

ID	KEYWORDS	DATA	COMMENTS
		in molten salt layer	
<i>Precipitation Extraction of Sulfate from SBW Simulant: Ref. I-3</i>			
I-0	Test description	<ul style="list-style-type: none"> • SBW simulant with 0.07 M sulfate • Ba(NO₃)₂ added at Ba/SO₄ ratios of 0.8, 1.0 and 1.2 • Temperature: Ambient and 60°C • Stir time: 5, 60, 120 min 	<p>Objectives:</p> <ul style="list-style-type: none"> • Evaluate effectiveness of sulfate removal from SBW by precipitation with Ba(NO₃)₂ • Measure settling and filtration rates of BaSO₄ precipitate • Assess effect of heating and stir time during precipitation on liquid/solid separation rates
I-1	Sulfate Precipitation Efficiency	<ul style="list-style-type: none"> • 57-87% sulfate precipitated • Sulfate precipitation increased with increasing Ba/SO₄ ratio and increasing temperature • Fraction of barium not precipitated increased with increasing Ba/SO₄ ratio and increasing temperature 	
I-2	Coprecipitation of strontium	<ul style="list-style-type: none"> • At ambient temperature, 2.5-3.2% of the Sr precipitated, with the amount decreasing with increasing Ba/SO₄ ratio • At 60°C, 1.7-8% of the Sr precipitated, with the amount decreasing with increasing Ba/SO₄ ratio 	
I-3	Coprecipitation of Pu	<ul style="list-style-type: none"> • At ambient temperature, 1.1-1.5% of the Pu precipitated, with the amount decreasing with increasing Ba/SO₄ ratio 	
I-4	Precipitate PSD	<ul style="list-style-type: none"> • Precipitate particle size range: ~0.4 - 150 µm, with the maximum particle volume occurring at 40-80 microns • The particle size increased with increased stir time • Increasing temperature slightly increased the fraction of larger particles 	<ul style="list-style-type: none"> • Samples stirred for 5, 60 and 120 minutes • Temperatures = ambient, 70°C
I-5	Settling Rate	<ul style="list-style-type: none"> • Majority of particles settle in 12 minutes; after settling overnight, the liquid appears completely clear • Non heated tests appear to have faster settling rates than heated tests • 78-84% of the sulfate precipitated, 10-20% of barium remained in solution • Velocity of the larger particles 0.029 cm/sec • Settling rate in wash water higher than in the waste solution 	<ul style="list-style-type: none"> • 4 tests: ambient temperature tests with 60 min and 120 min stir time; 5 min heating and 60 min stir time; 60 min heating and 120 min total stir time
I-6	Filtration Rate	<ul style="list-style-type: none"> • Filtration rate curves linear • Heated samples filtered faster than ambient temperature samples • 82-85% of the sulfate precipitated, 11-39% of the barium remained in solution 	<ul style="list-style-type: none"> • 4 tests: ambient temperature tests with 60 min and 120 min stir time; 5 min heating and 60 min stir time; 60 min heating and 120 min total stir time
I-7	Washing of	<ul style="list-style-type: none"> • Precipitates were washed and allowed to resettle. Solids from heated 	

ID	KEYWORDS	DATA	COMMENTS
	Precipitate	sample settled more rapidly. Both heated and non-heated solids resettled to a clear solution in 25 minutes.	
<i>MSE Performance Testing of NOxidizer: Ref. 14</i>			
J-0	Test description	<ul style="list-style-type: none"> • Optimization tests - NOxidizer inlet feed was air with ~50,000 ppm NOx - 16 tests based on factorial design of control parameters: T_{Quench} $\Delta T (=T_{\text{Quench}} - T_{\text{reoxidation}})$, $[O_2]_{\text{stack}}$ - Test conditions held 1 hour after reaching stable operation • Benzene injected (369 ppm) to determine DRE of organics • 1,500 µg/dscm Hg injected by vaporizing metal mercury and injecting Cl_2 to partially oxidize the mercury 	<p><u>Objectives:</u></p> <ul style="list-style-type: none"> - Measure NOx DRE using propane fuel - Measure NOxidizer THC emissions, with and without organic in the waste - Measure NOx speciation downstream of the reduction stage and reoxidation stage - Measure Hg speciation in NOxidizer effluent
J-1	NOx DRE	<ul style="list-style-type: none"> • 92-96% NOx DRE • 95-97% approach to equilibrium 	<ul style="list-style-type: none"> • 96% DRE achieved at 3 conditions: - $\Delta T = 200^\circ\text{F}$, $[O_2]_{\text{stack}} = 1.5\%$, $T_{\text{Quench}} = 1600^\circ\text{F}$ - $\Delta T = 50^\circ\text{F}$, $[O_2]_{\text{stack}} = 1.5\%$, $T_{\text{Quench}} = 1800^\circ\text{F}$ - $\Delta T = 50^\circ\text{F}$, $[O_2]_{\text{stack}} = 4\%$, $T_{\text{Quench}} = 1600^\circ\text{F}$
J-2	Organic DRE	<ul style="list-style-type: none"> • > 99.7% total hydrocarbon (THC) DRE 	<ul style="list-style-type: none"> • THC below instrument DL of 5 ppmv
J-3	NOx Speciation	<ul style="list-style-type: none"> • Reduction section outlet: [NO] drops from ~33 to 10 ppmv, NH_3 from ~68 to 10 ppmv, and NO_2 is below DL • Reoxidation section outlet: [NO] varies 400-600 ppmv; $[NO_2]$ varies 10-30 ppmv (spikes to 70 ppmv); $[HNO_3]$ drops to ~0 ppmv 	<ul style="list-style-type: none"> • Authors characterize NOx speciation tests as having had "limited success"
J-4	Hg Speciation	<ul style="list-style-type: none"> • Stack samples averaged 0.96 µg/dscm oxidized and 535 µg/dscm elemental Hg, indicating > 99.8% of the Hg was elemental 	<ul style="list-style-type: none"> • Injected Hg was 60% $HgCl_2$, 40% elemental • $[Hg]_{\text{stack}}$ indicated actual Hg in OG roughly 27 times what was injected—attributed to volatilization of Hg trapped in system from earlier test
<i>HLLWE Bottoms Sampling Results: Ref. 15</i>			
K-0	Study description	<ul style="list-style-type: none"> • Input to model was composition of WM-185, updated from analysis of samples taken in October 2001 	<p><u>Objectives:</u></p> <ul style="list-style-type: none"> • Compare ASPEN process model predictions for HLLWE bottoms composition with analytical measurements to identify areas where simulation can be improved
K-1	WM-181/184 Concentrate Composition	<ul style="list-style-type: none"> • HLLWE bottoms composition from evaporation of 2:1 WM-181:WM-184 blend: H^+ (2.75 M), Al (0.785 M), Cd (1.31E-3 M), Cl (2.95E-2 M), Cr (1.67E-3 M), F (6.44E-2 M), Hg (5.68E-4 M), NO_3^- (5.51 M), K (0.142 M), Na (1.74 M), U (3.14E-4) 	<ul style="list-style-type: none"> • ASPEN results are within 4% of analytical results for H^+ and Al, are 40-50% higher for most other species, and higher by a factor of 2-3 for Cd and Hg.

ID	KEYWORDS	DATA	COMMENTS
<i>Mercury Vapor-Liquid Equilibrium Testing: Ref. 16</i>			
L-0	Test description	<ul style="list-style-type: none"> • Scrub solution temperature: 70°C (prior test data was at 80 °C) • System pressure: ambient (~12.5 psia) • Scrub solution composition: HNO₃ (0.0-1.0 M), HgCl₂ (0.0001-0.1 M), Cl (0-0.3 M), total alkali metals (0-0.1 M) • Gas purge composition: N₂ (98-100%), NO₂ (0-2%) • Elemental Hg not used as reagent • Gas bubbled through liquid at 36 scc/min 	<p>Objectives:</p> <ul style="list-style-type: none"> • Measure concentrations of Hg in vapor and liquid that are approximately at equilibrium • Use vapor and liquid compositions that are representative of a production facility • Compare vapor/liquid partitioning to what has been measured previously
L-1	Hg VLE	<ul style="list-style-type: none"> • Log-log plots of y (vap mole frc Hg) vs x (liq mole frc Hg) are linear • For given x, y increases with temperature (70-80°C) • With 2% NO₂ in vapor phase and 1 M HNO₃ in liquid, slope of x-y line on log-log plot is higher than with no NO₂ • For HgCl₂ Raoult's Law gives y=(0.000019-0.000041)x @ T=70-80°C • For Hg Raoult's Law gives y=(0.000080-0.000145)x @ T=70-80°C • Slope of log-log plot of x vs y should be 1.0 based on Raoult's Law 	<ul style="list-style-type: none"> • Effect of acid and chloride concentration inconclusive • Prior tests at 80°C gave y \approx (0.002)x; these tests gave y \approx (0.004)x (no NO₂). Measured coefficient of x is 10-100 times Raoult's Law prediction (\Rightarrow more Hg in vapor) • Slope of log-log plot of x vs y slightly less than 1.0 with no NO₂ in gas • Slope of log-log plot of x vs y greater than 1.0 with 2% NO₂ in gas
<i>Virification Demonstration—RSM Test #2: Ref. 17</i>			
M-0	Test description	<ul style="list-style-type: none"> • Melter cross-section: 182 cm² • 20 wt% waste loading • Reductant concentrations: Sugar: 160-200 g/L; Glycolic: 336-420 g/L • Target RR range: 0.01-0.30 (one outlier data point at 0.9) 	<p>Objectives:</p> <ul style="list-style-type: none"> • Compare processing behavior with and without reductants (sugar, glycolic acid) in feed with respect to: <ul style="list-style-type: none"> - Emissions (S, Hg, halogens) - Glass redox ratio and nitrate reduction - Secondary waste stream compositions • Evaluate effectiveness of SBW-22 formulation in incorporating sulfur in glass • Test efficacy of real-time glass redox monitor • Quality of glass product
M-1	Sulfur behavior	<ul style="list-style-type: none"> • Salt layer not observed at sugar concentration of 160 g/L • Salt layer not observed with no reductant in feed 	<ul style="list-style-type: none"> • Salt left from previous test (RSM-1) was seen at beginning of test but eventually disappeared • Tests run for 10.3 and 10.6 hrs, respectively

ID	KEYWORDS	DATA	COMMENTS
M-2	Glass redox	<ul style="list-style-type: none"> Redox curves (RR vs reductant concentration) generated for SBW-9 not applicable to SBW-22 Glycolic acid not effective in controlling glass redox state 	<ul style="list-style-type: none"> Redox determinations for this test found to be invalid (after the fact) due to interference of vanadium used in formulation^b Glycolic acid behavior in RSM-2 not consistent with crucible tests
M-3	Foaming	<ul style="list-style-type: none"> Extensive foaming with no reductant and with excessive reductant Sugar and glycolic acid can control foaming 	<ul style="list-style-type: none"> Excessive reductant: Sugar > 200 g/L
M-4	Cold cap behavior	<ul style="list-style-type: none"> Thicker, more conventional cold cap with SBW-22 than with SBW-9 Thicker with no reductant in feed Thicker with glycolic acid than with sugar 	<ul style="list-style-type: none"> Insulating foam layer was observed as part of cold cap at high reductant concentrations in feed and with no reductant at all
M-5	Reductant type	<ul style="list-style-type: none"> Glycolic acid volatilizes more readily and reduces less Fe in glass 	<ul style="list-style-type: none"> Inconsistent with lab crucible tests
M-6	Feed Rheology	<ul style="list-style-type: none"> At 20% WL GFC additives neutralized feed slurry causing it to thicken, and required addition of HNO₃. This increased reductant requirement as well 	
M-7	Melt rate	<ul style="list-style-type: none"> Glass production rate for SBW-22 ~20% lower than for SBW-9 Area-specific glass rate higher than average for comparable melters 	
M-8	Slagging, plugging	<ul style="list-style-type: none"> Significant deposits inside FC No deposits found downstream of FC 	<ul style="list-style-type: none"> FC deposits not expected or typical, and were attributed to weld failure which affected flow pattern inside the FC
M-9	Melter offgas composition	<ul style="list-style-type: none"> 160-200 g/L sugar in feed: H₂(0.20%), NO(5.4%), CO(0.9%), CO₂(8.8%), N₂O(0.3%), THC(0.07%), NO₂(1.05%) 280-392 g/L glycolic acid in feed: H₂(0.37%), NO(5.5%), CO(0.2%), CO₂(9.6%), N₂O(0.43%), THC(0.054%), NO₂(0.65%) No reductant in feed: H₂(NM), NO(7.4%), CO(NM), CO₂(1.2%), N₂O(NM), THC(0.10%), NO₂(4.7%) CO and H₂ well below flammability limits 	<ul style="list-style-type: none"> %s at left are arithmetic means of average concentrations in Table 6.2 of report, with outliers excluded Table 6.2 values were back-calculated from measured gas composition at FC exit, assuming no melter are inleakage Without reductants NO₂ concentration is higher by factor of 4-7x Glycolic acid appears more effective in reducing NO₂ (factor of ~2) SO₂ not detected above ~8 ppm (SO₂ measurements discussed in Ref. 27) Gas compositions similar to RSM-1 ~73% of NO₃- is reduced

^b Reported by Dr. Duane Siemer and documented in a Lotus Notes memorandum from Keith Perry to Lance Lauerhass on 11/7/01 titled "Redox measurement discovery".

ID	KEYWORDS	DATA	COMMENTS
M-10	Melter DF (overall)	<ul style="list-style-type: none"> Sugar: 34 Glycolic acid: 204 No reductant: 109 	<ul style="list-style-type: none"> Higher DF with glycolic acid and with no reductant attributed to better-developed cold caps compared to sugar
M-11	Melter DFs (semivolatiles)	<ul style="list-style-type: none"> Range: 9-817 Cl(NM), F(NM) Element-specific: B(18-107), Ba(13-72), Ca(82-817), Cd(52-104), Cr(34-178), Cs(9-49), K(14-85), Li(46-306), Na(17-100), Pb(37-576), Ru(20-59), Sr (NM), Zn(30-142) 	<ul style="list-style-type: none"> From data in Table 6.13 DF=(feed mass)/(feed mass not in glass) Reported DFs generally much higher than for RSM-1 For additional information see Attachment 2
M-12	Melter DFs (volatiles)	<ul style="list-style-type: none"> Range: 1-1220 Element-specific: P(104-1,220), S(15-50), Hg(1-6), Cl(1-14), F(1-23), I(1-2) Only species found in gas phase measurements were B, halogens, and Hg 	<ul style="list-style-type: none"> HCl/Cl₂ levels in OG were 9-90 ppm (see Ref. 27) Halogens measured in gas phase but not in particulates--DFs are therefore upper bounds
M-13	Melter DFs (nonvolatiles)	<ul style="list-style-type: none"> Range: 58-1560 Element-specific: Fe(73-1560), Si(NM), Al(58-461), Ce(NM) 	"
M-14	Melter DFs (others)	<ul style="list-style-type: none"> Element-specific: Cu(65-203), Mn(35-1,460), Mo(29-172), Mg(74-12000), Gd(70-1870) 	"
M-15	Fate of mercury	<ul style="list-style-type: none"> Measured Hg speciation in melter offgas: <ul style="list-style-type: none"> - Elemental: 87-98% (with reductant), 64% (without reductant) - Particulate: 0.1-0.3% (with reductant), 5.1% (without reductant) 27% of Hg from feed was measured in scrub liquids and UDS, 46% was found in acid washes of HEME mesh (post-test) 97% of Hg removed by [EVS + HEME] Hg detected in vapor space of scrub samples using hand-held analyzer (however, analyzer also responds to nitric acid-generated vapors) No Hg detected in HEME OG exhaust 	<ul style="list-style-type: none"> High indicated Hg scrubbing efficiency not expected from high fraction of elemental Hg 97% scrubbing efficiency is based on comparing CEM Hg measurements upstream and downstream of scrub system for a single condition (with glycolic acid as reductant) Overall ~73% of the Hg fed to the melter was recovered in the scrub system
M-16	Sulfur partitioning	<ul style="list-style-type: none"> %Sulfur as SO₄²⁻ in scrub liquors: 83% (sugar), 31% (glycolic acid) Noncontiguous pools of molten salt were observed on glass melt early in run but steadily decreased as test progressed Not all S was retained by glass--5-10% of feed S caught by scrub system^c 0.7-1.05 wt% SO₃ measured in glass (all conditions) Glass wt% SO₃ appears to have increased as feed concentration of sugar decreased during first three days of testing 	<ul style="list-style-type: none"> Molten salt observed was attributed to residual glass present in melter from RSM-1 test Trend of higher wt% SO₃ in glass with lower sugar concentration in feed consistent with prior results (entry H-2, above) Above trend reversed after sugar was re-introduced as reductant following 24 hrs of feed with glycolic acid as reductant

^c An error in a spreadsheet was discovered after release of the final report. The error impacts the reported proportions of feed elements collected in scrub solutions (Figure 6.8 of Ref. 17) and waste stream-derived melter DFs (Table 6.19 of Ref. 17) for all species except Hg, F, Cl, and I, which were reported correctly. Fractional retentions in scrub solutions reported in Attachment 2 (this report) were generated independently, and are believed to be correct. Sulfur retention of 5-10% is from Attachment 2.

ID	KEYWORDS	DATA	COMMENTS
M-17	Scrubbing performance	<ul style="list-style-type: none"> Majority of species extracted from OG were collected in HEME rather than EVS Main species in scrub liquids: Na and C Main species in UDS from scrub samples: Fe and Si 	<ul style="list-style-type: none"> See Table 6.15 of report for comparison of EVS and HEME
M-18	Real-time glass redox monitoring	<ul style="list-style-type: none"> Use of platinum electrodes sheathed in alumina tubes and purged with O₂ and Ar gases to give real-time redox measurement failed due to molten glass dissolving sheathing material 	<ul style="list-style-type: none"> Concept may still be viable if corrosion-resistant sheathing material can be found
M-19	Glass durability	<ul style="list-style-type: none"> Normalized PCT release rates as % of EA Glass values: B(<7.8%), Li(<31%), Na(<17%), Si(<42.7%) TCLP concentration as % of UTS: Ni(<30%), Ba(<0.23%), Pb(<13%), Cr(<8.3%), Cd(<18%), Hg(<80%), 	<ul style="list-style-type: none"> Leaching performance of all glass samples tested exceeded that of EA glass and UTS requirements Nickel data outliers excluded from results
<i>Rheology Testing of SBW: Ref. 18</i>			
N-0	Test description	<ul style="list-style-type: none"> Silica sources tested: F-110 Glassmakers sand (~110 mesh), -240 mesh, -325 mesh, and diatomaceous earth Settling rates compared by measuring volume of liquid containing solids in a graduated after a prescribed settling time Feed stability determined observing changes to solution rheology, color, consistency, and foaming over 1 week period Feed viscosities measured in an M-5 Haake rotational viscometer for shear rates from 0 to 1000 sec⁻¹ Feeds were dried in a purged, stirred 2-liter reaction vessel on hotplate. Acid concentrations of 1.1 and 3.0 M were tested and OG behavior was observed. NO, NO₂, N₂O, CO₂, CO, H₂ concentrations measured. 	<p>Objectives:</p> <ul style="list-style-type: none"> Screen silica sources for melter feed based on settling rates observed in lab tests Evaluate stability of melter feeds made with different silica sources Measure viscosity of melter feed at various strain rates to identify non-newtonian behavior Measure the temperature at which observable sugar denaturation begins in SBW feed simulant containing sugar
N-1	Feed Settling	<ul style="list-style-type: none"> Diatomaceous earth as a silica source resulted in some foaming The larger the silica particle size, the faster particles settled and the more solids accumulated in the bottom of the cylinder No evidence of gelling or caking; settled solids easily resuspended Bubbles in some tests may have indicated a reaction with sugar 	<ul style="list-style-type: none"> Surrogate SBW was thoroughly mixed with glass forming components (and sugar in some tests), then poured into a 100-ml graduated cylinder
N-2	Feed Stability	<ul style="list-style-type: none"> No gelling or caking was observed; no changes occurred in the rheology of the mixture or the behavior of the solids Excessive foaming was seen in the tests with F-110 Glassmaker's Sand and silica without iron oxide and with 325 mesh silica without iron oxide. Less foaming was seen in the same tests with iron oxide. In all tests, the Teflon coating on the stir bar either wore off and the stir bar partially of totally dissolved, or the stir bar wore a hole in the beaker Significant pH change noted in samples mixtures where stir bar dissolved Feed mixtures generally changed color (becoming darker) over time 	<ul style="list-style-type: none"> Six surrogate feed mixtures prepared and mechanically stirred in a beaker for at least 1 week Silica sand slurries are quite erosive

ID	KEYWORDS	DATA	COMMENTS
N-3	Feed Rheology	<ul style="list-style-type: none"> Viscosity of liquid SBW simulant alone = 3.2 cP SBW simulant appears to be slightly non-Newtonian Viscosity of SBW melter feed 5.2 cP SBW melter feed is Newtonian 	<ul style="list-style-type: none"> Plots of shear stress as function of Shear rate for SBW simulant and melter feed given
N-4	Sugar Reaction	<ul style="list-style-type: none"> No reaction detected at ambient temperature Reaction between sugar and surrogate waste observed to begin at 50-80°C based on detection of NO_x, CO and CO₂ in purge gas 	<ul style="list-style-type: none"> Simulated melter feed stirred in purged vessel, purge gas analyzed for NO_x, test run for several days
<i>Scrubber Evaluations: Ref. 19</i>			
O-0	Study description	<ul style="list-style-type: none"> “Constituents of primary concern” (COPCs) in the OG identified. Removal requirements for COPCs and other process process requirements discussed Alternative OG treatment systems described and discussed Baseline process simulated with ASPEN Plus; capabilities and limitations of ASPEN Plus model identified 	<p>Objectives:</p> <ul style="list-style-type: none"> Identify process and operating objectives for the OG treatment system Evaluate existing mixed waste process OG treatment systems and unit operations Select baseline OG treatment system Develop ASPEN model to model baseline process and evaluate alternatives
O-1	Baseline Process	<ul style="list-style-type: none"> Baseline OG treatment process consists of film cooler, acid quench, venturi scrubber, HEME, heater, HEPA filters, NOx oxidizer, caustic quench, demister, reheater, GAC bed, and final HEPA filters 	
<i>Secondary Waste Considerations: Ref. 20</i>			
P-0	Study description	<ul style="list-style-type: none"> Three disposal sites considered: Hanford, NTS, Envirocare RWMC not considered because not a long term mixed waste disposal site Considered (a) radionuclide concentration limits, (b) Hg content limits Considered five secondary waste streams: (a) Grouted scrubber blowdown, (b) Spent Cs IX resin, (c) HEPA filters (primary), (d) HEPA filters (secondary), (e) Spent GAC Rad dosage calculated using MicroShield computer code Grout meets Hanford Category 3 and NRC Class C low-level waste limits Since Hg content exceeds 260 mg/kg a demonstration of equivalency to BDAT (amalgamation) would be needed for disposal Leaching tests show grout containing blast furnace slag will pass TCLP with Hg levels up to 1910 mg/kg Expected contact dose from waste in 55-gal drums < 1 mR/hr Recommended combining acid and caustic scrub purges prior to IX, and removing both cesium and strontium by IX 	<p>Objectives:</p> <ul style="list-style-type: none"> Estimate key properties of projected secondary wastes from vitrification Evaluate impacts of estimated properties on disposal options for treated secondary wastes
P-1	Scrubber Blowdown Grout Disposal		<ul style="list-style-type: none"> Removal of ⁹⁰Sr from the acid scrub recommended to reduce the classification of the waste to Hanford Category 1 or NRC Class A Combining of caustic and acid scrubs would reduce pH and improve efficiency of Sr removal

ID	KEYWORDS	DATA	COMMENTS
P-2	Ion Exchange Media Disposal	<ul style="list-style-type: none"> • Could be recycled to the melter and disposed with glass • Could be disposed directly in high integrity containers at Hanford as remote handled Category 3 waste 	<ul style="list-style-type: none"> • Recycle of the spent IX sorbent to the melter is recommended to avoid a secondary waste stream • Extraction of Cs-137 only required if contact-handled grouting process is deemed necessary
P-3	HEPA Filter Disposal	<ul style="list-style-type: none"> • Spent primary HEPA filters qualify as Hanford Category 3 or NRC Class C waste; however, Hg in the filters must be rendered non-leachable before they would be accepted by disposal sites • Spent secondary HEPA filters could be disposed at Envirocare • 816 HEPA HVAC filters must be disposed at facility closure 	<ul style="list-style-type: none"> • Using washable filters as primary HEPAs is recommended to eliminate a secondary waste • Upstream removal of mercury is an alternative method to avoid leachable Hg on the primary HEPA filters
P-4	Spent GAC Disposal	<ul style="list-style-type: none"> • Spent carbon could be disposed at Hanford, NTS, or Envirocare • Leaching tests needed to determine maximum allowable Hg loading on carbon and confirm long-term retention of Hg • Demonstration of equivalency to BDAT (amalgamation) needed 	<ul style="list-style-type: none"> • Consideration should be given to upstream Hg oxidation, scrubbing and removal from scrub to reduce or eliminate spent GAC waste • Demonstration of equivalency likely to require approval of disposal sites and regulators
P-5	Envirocare WAC	<ul style="list-style-type: none"> • Listed waste: Accepts F001, F002, F003, and U134 (all in SBW) • RCRA metals: Accepts waste passing TCLP • No free liquids • No requirement for compressive strength for solid wastes • 55-gal drum container OK for most waste classes • Cat. 1 radionuclide concentration limits (Ci/m³): ⁹⁰Sr(0.016), ⁹⁹Tc(0.023), ¹²⁹I(0.0085), ¹³⁷Cs(0.0055), ²⁴¹Pu(0.061), ²³⁸Pu(0.0047), ²³⁹Pu(0.0019), ²⁴⁰Pu(0.0019), ²⁴¹Am(0.021) • Cat. 3 radionuclide concentration limits (Ci/m³): ⁹⁰Sr(54000), ⁹⁹Tc(5), ¹²⁹I(1.8), ¹³⁷Cs(12000), ²⁴¹Pu(25), ²³⁸Pu(24), ²³⁹Pu(0.42), ²⁴⁰Pu(0.43), ²⁴¹Am(0.85) 	<ul style="list-style-type: none"> • See page 3 of Ref. 20 • Envirocare has only one LLW category for purposes of defining maximum radionuclide concentrations in waste
P-6	Hanford WAC	<ul style="list-style-type: none"> • Listed waste: Accepts F001, F002, F003, but NOT U134 (all in SBW) • RCRA metals: Acceptance based on concentration in waste, not in TCLP leachates • No free liquids • No requirement for compressive strength for solid wastes • 55-gal drum container OK for most waste classes • High-activity wastes must be packaged in a HIC • Cat. 1 radionuclide concentration limits (Ci/m³): ⁹⁰Sr(0.016), ⁹⁹Tc(0.023), ¹²⁹I(0.0085), ¹³⁷Cs(0.0055), ²⁴¹Pu(0.061), ²³⁸Pu(0.0047), ²³⁹Pu(0.0019), ²⁴⁰Pu(0.0019), ²⁴¹Am(0.021) • Cat. 3 radionuclide concentration limits (Ci/m³): ⁹⁰Sr(54000), ⁹⁹Tc(5), ¹²⁹I(1.8), ¹³⁷Cs(12000), ²⁴¹Pu(25), ²³⁸Pu(24) 	<ul style="list-style-type: none"> • See page 3 of Ref. 20 • Acceptance level for Hg concentration not specified

ID	KEYWORDS	DATA	COMMENTS
P-7	NTS WAC	<p>$^{239}\text{Pu}(0.42)$, $^{240}\text{Pu}(0.43)$, $^{241}\text{Am}(0.85)$</p> <ul style="list-style-type: none"> Listed waste: Accepts F001, F002, F003, but NOT U134 (all in SBW) RCRA metals: Accepts waste passing TCLP No free liquids No requirement for compressive strength for solid wastes 55-gal drum container OK for most waste classes Cat. A radionuclide concentration limits (Ci/m^3): $^{90}\text{Sr}(0.04)$, $^{99}\text{Tc}(0.3)$, $^{129}\text{I}(0.008)$, $^{137}\text{Cs}(1)$, $^{241}\text{Pu}(10)$, $^{238}\text{Pu}(10)$, $^{239}\text{Pu}(10)$, $^{240}\text{Pu}(10)$, $^{241}\text{Am}(10)$ Cat. C radionuclide concentration limits (Ci/m^3): $^{90}\text{Sr}(7000)$, $^{99}\text{Tc}(3)$, $^{129}\text{I}(0.08)$, $^{137}\text{Cs}(4600)$, $^{241}\text{Pu}(3500)$, $^{238}\text{Pu}(100)$, $^{239}\text{Pu}(100)$, $^{240}\text{Pu}(100)$, $^{241}\text{Am}(100)$ 	<ul style="list-style-type: none"> See page 3 of Ref. 20 Radionuclide concentration limits based on NRC guidelines for Cat. A and Cat. C LLW
<i>Modeling of NOx Destruction Options: Ref. 21</i>			
Q-0	Study description	<ul style="list-style-type: none"> 3 kinetic models developed using the Chemkin code: <ul style="list-style-type: none"> Staged combustion as three plug-flow reactors in series SNCR as a single plug-flow reactor High-temperature NO_x abatement as a single plug-flow reactor Models include 88 chemical species and 546 reactions 	<p>Objectives:</p> <ul style="list-style-type: none"> Develop computational models for NO_x abatement methods suitable for integration with steady state mass balance calculations and for inclusion in CFD code Exercise models to assess feasibility of these NO_x abatement methods
Q-1	Staged combustion	<ul style="list-style-type: none"> Model input/basis: <ul style="list-style-type: none"> $6.5^\circ\text{D} \times 23^\circ\text{L}$ reduction stage / $6.5^\circ\text{D} \times 5^\circ\text{L}$ quench / $5.8^\circ\text{D} \times 21^\circ\text{L}$ oxidation stage Inlet gas (mol %): $\text{CO}_2(2.09)$, $\text{CO}(0.29)$, $\text{H}_2\text{O}(27.4)$, $\text{N}_2(54.08)$, $\text{NO}(0.88)$, $\text{NO}_2(0.88)$, $\text{O}_2(12.8)$, $\text{Ar}(0.11)$, $\text{H}_2(0.47)$; Flows: Inlet gas($3300 \text{ m}^3/\text{hr}$), natural gas fuel($15.4 \text{ lb}/\text{min}$), air to reduction($1898 \text{ scfm}$), air to reoxidation($596 \text{ scfm}$), quench water to reduction(4 gpm), quench water to quench section(6 gpm) Overall heat transfer coefficient: $2500 \text{ erg}/\text{cm}^2\text{-sec-K}$ for each stage Model Results (reduction/quench/reoxidation stages) <ul style="list-style-type: none"> Residence time: $1.24 / 0.41 / 1.21$ seconds Exit velocity: $13 / 12 / 17 \text{ ft}/\text{sec}$ Outlet temperature: $2314 / 1672 / 1749^\circ\text{F}$ Outlet flow: $8231 / 9998 / 10,908 \text{ scfm}$ Outlet composition: $\text{CO}(23 \text{ ppbv})$, $\text{NO}(528 \text{ ppmv})$, $\text{NO}_2(2 \text{ ppmv})$, $\text{H}_2(111 \text{ ppbv})$, $\text{CO}_2(6.4 \text{ mol}\%)$, $\text{H}_2\text{O}(46.5 \text{ mol}\%)$, $\text{N}_2(45.7 \text{ mol}\%)$, $\text{O}_2(0.9 \text{ mol}\%)$, $\text{Ar}(0.4 \text{ mol}\%)$ 	<ul style="list-style-type: none"> Ratio of outlet to inlet flowrate = 6.0 Additional model results in Table 5 and Figure 7 of report

ID	KEYWORDS	DATA	COMMENTS
Q-2	SNCR	<ul style="list-style-type: none"> Model input/basis: <ul style="list-style-type: none"> SNCR located directly downstream of melter SNCR dimensions: 5'D x 10' L Inlet gas (mol %) CO₂(5.93), CO(0.81), H₂O(54.5), N₂(25.4), NO(2.48), NO₂(2.48), O₂(5.45), Ar(0.32), H₂(2.63) Flows: Inlet gas(1100 m³/hr) Reductant: ammonia Isothermal reaction (heat of reaction and heat losses neglected) Temperature: 800-1100°C (10°C increments) NH₃/NOx ratios: 1.0-1.4 Pressure: 1 atm 279 cases (31 temperatures x 9 NH₃/NOx ratios) <u>Model Results</u> <ul style="list-style-type: none"> Optimum operating temperature range is 850°C inlet and outlet ≤ 950°C Optimum NH₃/NOx ratio: 1.2 SNCR is viable option for NOx abatement in vitrification OG 	<ul style="list-style-type: none"> Additional model results shown on Figures 8-11 of report
Q-3	High-Temperature NOx abatement	<ul style="list-style-type: none"> Model input same as SNCR except for temperature range and reductant Without reductant, 3950 ppm NOx is theoretical minimum achievable concentration, achieved at a temperature of 1840°C Addition of H₂ as reductant does not significantly reduce NOx CH₄ enhances NOx reduction, but not as much as NH₃ Use of NH₃ makes this case identical to the SNCR Without a reductant, high temperature NOx abatement is not feasible 	<ul style="list-style-type: none"> Thermodynamic models were run to determine theoretical limits of high temperature NOx abatement Additional model results are shown in Figures 12-13 in the report
<i>Evaluation of Wet Vs Dry Feed: Ref. 22</i>			
R-0	Study description	<ul style="list-style-type: none"> Qualitative risks and disadvantages of partial evaporation compared to full evaporation disqualify it from detailed analysis and further study Preliminary design for a dry-feed melter system generated Cost analyses for dry and wet feed systems compared 	<u>Objectives:</u> <ul style="list-style-type: none"> Evaluate advantages/disadvantages of partial and full evaporation of feed to SBW melter Compare with slurry feed to melter
R-1	Incremental costs for dry feed	<ul style="list-style-type: none"> \$5 million savings (out of \$738 million) in capital cost for the scheme in which SBW is evaporated to dryness and fed dry to the melter The dry feed scheme would reduce the processing schedule by 30%, or for the same schedule as the baseline flowsheet, would result in additional capital cost savings The dry feed scheme has an incremental \$7 million (out of \$612 million) life cycle cost 	<ul style="list-style-type: none"> The cost estimates were derived from adjustments of the FY-2000 SBW Vitrification Feasibility Study (an upstream NOx oxidizer flow scheme) and are based on comparing a scheme in which SBW is evaporated to dryness in a wiped film evaporator rather than fed as liquid. The only capital cost differences were from feed system equipment and the addition of a film cooler for the dry feed case. Savings in

ID	KEYWORDS	DATA	COMMENTS
			OG equipment due to reduced flowrates are expected to be significant but were not included. The life cycle cost increment was based on reducing the baseline 2-yr schedule to 1.4 years.
R-2	Partial Evaporation	<ul style="list-style-type: none"> The study concluded no clear benefit to partial evaporation of SBW feed 	
<i>Feasibility Study for Off Gas Treatment from SBW Virrification: Ref. 23</i>			
S-0	Study description	<ul style="list-style-type: none"> Steps in evaluations: <ul style="list-style-type: none"> Identify contaminants of concern to be controlled by the OG system Develop baseline flow diagram and mass balance Review OG treatment technologies used at other DOE sites Considerations in assessment: <ul style="list-style-type: none"> Recycle of secondary streams to melter must accommodate QC required for acceptance of glass at Yucca Mtn Secondary wastes must be treated for disposal Only commercially available unit operations considered Minimize equipment that must be operated remotely in a hot cell 	<u>Objectives:</u> <ul style="list-style-type: none"> Evaluate possible OG treatment systems for the SBW vitrification process Propose a baseline design for the OG system
S-1	Baseline Process	<ul style="list-style-type: none"> Process Requirements identified <ul style="list-style-type: none"> MACT emissions limits Secondary waste disposal requirements Process functional and operational requirements NOx abatement to satisfy five identified requirements Baseline Process selected Treatment systems at other sites discussed NOx abatement technologies evaluated Process through which integrated baseline was selected described 	<ul style="list-style-type: none"> Considerable detail in evaluation of NOx abatement technologies in main report and Appendices Baseline process selected through a VE session on April 17-19, 2001 wherein six alternative flowsheets were evaluated using criteria developed by the participants Primary difference between schemes evaluated for baseline process: Method and/or location in the process of NOx abatement
S-2	NOxidizer data	<ul style="list-style-type: none"> Sizing and expected performance for NOxidizer (calculations from vendor based on feed composition and rate from INEEL) are provided Basis design: Reducing zone/Quench zone/Reoxidation zone Temperature out, °F: 2400/1665/1810 Fuel, MM Btu/hr: 18-20/0/0 Air, scfm: 1898/0/596 Quench water, gpm: 4/6/0 ID x OD x L, ft: 6.5x7.6x23 / 6.5x7.6x5-7 / 5.8x7.2x21 	<ul style="list-style-type: none"> See App. B of EDF-IWVF-004, which is in App. B of the Feasibility Study Air rate to reduction chamber recommended by vendor is 7.2 times higher than was assumed in mass balance (Reason: Needed for burner stability per price quotation included in report)
S-3	Mersorb performance	<ul style="list-style-type: none"> Expected performance of carbon (test data from vendor) <ul style="list-style-type: none"> For Hg in air at conc of 32 mg/m³, 30°C, 25 mm x 152 mm bed, 3 mm 	<ul style="list-style-type: none"> Contained in Appendix H of report

ID	KEYWORDS	DATA	COMMENTS
		<p>pellets, 1 atm, and 3 ft/sec, Hg removal efficiency versus time for several residence times show that a residence time of 10 seconds is required to maintain high (~100%) removal</p> <ul style="list-style-type: none"> - Hg capacity: based on a 407 day test of Hg in air, capacity of the bed divided into 6 segments was 23/19/15/15/15/0.3 g Hg/g carbon - Wt loss of absorbent vs temperature: no weight loss until 225°C, ~0.6% at 250°C, ~4% at 275°C 	
<i>SBW Feed Composition to Treatment Process: Ref. 24</i>			
T-0		<ul style="list-style-type: none"> • Data compiled from tank farm management plans, recent tank farm liquid and solids sampling reports, INEEL reports on NGLW projections • Projections based on two SBW treatment scenarios: <ul style="list-style-type: none"> - Processing of existing SBW from 2011-2013 and NGLW during 2016 - Processing all tank farm SBW and NGLW from 2016-2018 • Report to be updated as tank farm management plans change and as new data on tank farm contents becomes available 	<u>Objectives:</u> <ul style="list-style-type: none"> • Compile available data on volumes and compositions of liquids and solids in SBW wastes in INTEC tank farm facility • Incorporate projected volumes and compositions of newly generated liquid waste to provide informed estimates of waste volumes and compositions to be treated in the SBW vitrification facility
T-1	SBW Feed Composition	<ul style="list-style-type: none"> • Compositions are given in the report for <ul style="list-style-type: none"> - Liquid waste in each waste tank in the existing Tank Farm - Liquid waste in each waste tank in a new Tank Farm or if blended in the Vitrification Facility feed tanks <ul style="list-style-type: none"> - New Tank Farm SBW waste converted to oxides - Tank heel solids - Entrained or suspended solids - Newly generated liquid waste by year 	<ul style="list-style-type: none"> • This report is an update of a October 2000 report containing projected vitrification plant feed compositions. The revised compositions were based on updated waste analyses and INTEC Tank Farm management plans current as of July, 2001. Since that time, plans for construction of a new tank farm have been melded into the vitrification facility project, thus delaying their construction.
<i>WM-180 SBW Simulant Composition: Ref. 25</i>			
U-0	Task description	<ul style="list-style-type: none"> • Updated composition of waste in Tank WM-180 is given in the report, for both liquid and solids • Solids obtained were both entrained and suspended solids. The proportions of each were not determined • Reagents and procedure for preparing WM-180 simulant are given • Radionuclide isotopic concentrations are given 	<u>Objectives:</u> <ul style="list-style-type: none"> • Define the composition of a reasonable cold simulant for WM-180 liquid waste • Generate a "recipe" for making the simulant • Tank WM-180 was sampled and analyzed in late 2000 after transfer to tank NCC-102 in the NWCF.
U-1	WM-180 Composition	<ul style="list-style-type: none"> • Hydrated water in UDS amounts to about 24% of the mass of the solids • Estimated dried solids composition (wt %): 28.9% NaNO₃; 26.6% 	<ul style="list-style-type: none"> • Analytical results for samples were adjusted for residual wastes pre-existing in sampling

ID	KEYWORDS	DATA	COMMENTS
		Al(NO ₃) ₃ ; 12.77% K ₃ H ₆ Al ₅ (PO ₄) ₈ ; 5.43% FePO ₄ ; 4.48% SiO ₂ ; 3.74% ZrO ₂ ; 2.85% AlPO ₄ ; 1.8% Al ₂ (SO ₄) ₃ ; 1.03 % Ca(OH) ₂ ; Ca ₃ (PO ₄) ₂ ; 0.269% SnO ₂ ; 0.0053% Ag ₂ O; 0.0048% Sb ₂ O ₃	tank (NCC-102) <ul style="list-style-type: none"> Thermodynamic modeling and solids sample analyses used to estimate solids composition
U-2	Tank solids particle size distribution	<ul style="list-style-type: none"> Bell-shaped distribution, mean particle size 10 µm Most particles 2-50 µm 	<ul style="list-style-type: none"> Measured with Horiba LA-300 laser scattering PSD analyzer
U-3	Solids crystallinity	<ul style="list-style-type: none"> No discernable peaks in XRD analysis. Solids may be amorphous 	<ul style="list-style-type: none"> Alternative explanation is that there were insufficient solids to elicit XRD response
<i>"Worst Case" SBW Simulant Composition: Ref. 26</i>			
V-0	Task description		<p>Objectives:</p> <ul style="list-style-type: none"> Define criteria for a "worst case" simulant Estimate "worst case" concentrations of problematic species in feed per the criteria defined Generate a "recipe" for making the simulant Limiting species should be at their maximum concentration in order for the waste to be "worst case"
V-1	Worst Case Criteria	<ul style="list-style-type: none"> Criteria for "worst case": Concentrations of all species deemed to (a) limit waste loading, and/or (b) maximize the required level of OG treatment should be upper bounds for expected concentrations in melter feeds under expected SBW treatment scenarios Limiting species relative to glass formulation and glass properties: S, halides, P, noble metals (Pd, Ru, Rh, Ag), Cr, Al Limiting species relative to OG treatment: Acid, total nitrate, Hg, I, Cs, Tc, Sb and Te Reagents and procedure to be used in preparing the worst case simulant are given in the report 	
V-2	Worst Case Composition	<ul style="list-style-type: none"> Liquid waste composition (Molar): H⁺(3.56), Al(0.696), Sb(6.7E-5), Br(7.4E-6), Ca(7.3E-2), Cs(2.2E-5), Cl(3.15E-2), Cr(3.5E-3), F(0.12), I(3.7E-6), Hg(7.0E-3), Nd(3.8E-5), Pd(2.5E-5), PO₄³⁻(1.4E-2), Rh(4.6E-6), Ru(1.3E-4), Ag(2.2E-5), SO₄²⁻(7.3E-2), Tc(1.3E-5), Te(3.7E-6), U/Ce(6.5E-4) 	<ul style="list-style-type: none"> Limiting concentrations for Ca and Cr based on waste projected to be in WM-189; H⁺, Nd, Br, Cs, F, I, Hg, Rh, Ag, Tc, Te, and U on WM-188; and the other species on WM-180 Some species [notably Cs and Rh (as surrogate for Tc)] should be increased in surrogates above waste concentrations to ensure detectability in test samples
V-4	Effect of UDS on waste composition	<ul style="list-style-type: none"> Species present in high concentration in UDS that have a significant effect on the waste composition are B, Fe, Nb, Ni, Si, Zr, Cl, F, PO₄, SO₄ assuming an estimated average solids composition for all the waste tanks. Based on the WM-180 UDS composition alone, the list is reduced to Fe, Nb, Si, Zr and PO₄ 	<ul style="list-style-type: none"> "Significant" is defined as a molar concentration ≥5% of the liquid concentration in the worst case simulant (assuming 5 g/liter UDS). The most significant contributions are Si (conc. in solids = 2,700,000% of conc. in

ID	KEYWORDS	DATA	COMMENTS
			worst case liquid), Nb (589%), B (126%), PO ₄ (54%), Zr (44%), and SO ₄ (16%) <ul style="list-style-type: none"> Threshold for significance should be redefined relative to liquid concentrations including frit
V-5	Surrogates	<ul style="list-style-type: none"> Use Ce for U and Pu Use Nd for Am Use U for Np 	<ul style="list-style-type: none"> Recommendations for Am, Pu, and Np based on phase behavior in glass Basis of Ce for U not specified
V-6	Free HF in waste	<ul style="list-style-type: none"> Upper limit on free HF concentration in SBW tank wastes is 5.65×10^{-4} M Free HF concentration requiring additives [Al(NO₃)₃ or H₃BO₃] to control corrosion is 0.055 M (corrosion of 304L SS at 35°C is 2 mils/yr at this HF concentration) 	<ul style="list-style-type: none"> Results were based on worst case SBW simulant composition from Ref. 26 See Ref. 29 for details

2.2 DBEs Impacted by 2001 Tasks

Many of the FY 2001 tasks generated results which impacted some DBEs. Table 2-2 identifies which DBEs were so-impacted. Columns in the table represent all 22 tasks completed in FY 2001 and column headings give references to the respective reports which describe them. Rows in the table represent the DBEs. For each DBE listed the column entries below the DBE name indicate which tasks impacted that DBE and provides the row IDs from Table 2-1 which contain the relevant information.

Table 2-2: Impacts of FY-2001 development tasks to DBEs.

D B E	CETL #1 Ref. 5	Hg/GAC tests Ref. 6	Melt rate tests Ref. 7	Rad crucible Ref. 8	CVS Ref. 9	Off-gas systems Ref. 10	RSM#1 Ref. 11	S Partitioning Ref. 12	Sulfate sep'n Ref. 13	MSE NOx tests Ref. 14	HLLWE Ref. 15	VLE tests Ref. 16	RSM#2 Ref. 17	Rheology Ref. 18	Scrubber Ops Ref. 19	2nd Wastes Ref. 20	NOx model Ref. 21	Wet /dry feed Ref. 22	Off-gas FS Ref. 23	Feed char Ref. 24	WM-180 Ref. 25	Worst case Ref. 26
3							G-4			Detection of sulfate salt layer in the melter												
16										Submicron particle size distribution out film cooler												
	A-17																					
17										Nitrate destruction chemistry in melter												
	A-14												M-9									
18										Melter chemistry												
	A-6 A-14							G-13														
19										Target concentration of total dissolved solids (TDS) in melter feed												
								G-1						N-1 N-2 N-3				R-2				
20										Reductant selection and concentration in melter feed												
			C-5		E-5 E-6		G-3 G-14						M-1 M-5									
22										Pumping behavior of melter feed slurry												
													M-6	N-1 N-2 N-3								
24										Partition factors for feed species in melter												
	A-4 A-5 A-6 A-7 A-8						G-5 G-6 G-7 G-8						M-11 M-12 M-13 M-14 M-15									
25										Iodine-129 concentration in SBW feed												
																					T-1	
27										Partitioning of acid gases and Hg during quench/scrubbing operations												
												L-1	M-12 M-15									
29										Carryover of melter feed to offgas												
	A-4						G-7						M-10 M-12									
30										Representativeness of cold simulants in melter tests												
				D-1			G-3															
31										Composition envelope for acceptable glass feed												
	A-3				E-1																	

D B E	CETL #1 Ref. 5	Hg/GAC tests Ref. 6	Melt rate tests Ref. 7	Rad crucible Ref. 8	CVS Ref. 9 E-2 E-3 E-4 E-5	Off-gas systems Ref. 10	RSM#1 Ref. 11	S Partitioning Ref. 12	Sulfate sep'n Ref. 13	MSE NOx tests Ref. 14	HILWE Ref. 15	VLE tests Ref. 16	RSM#2 Ref. 17	Rheology Ref. 18	Scrubber Ops Ref. 19	2nd Wastes Ref. 20	NOx model Ref. 21	Wet /dry feed Ref. 22	Off-gas FS Ref. 23	Feed char Ref. 24	WM-180 Ref. 25	Worst case Ref. 26
	A-11 A-12																					
34	Frit vs GFC in melter feed																					
													M-6	N-1 N-2 N-3								
35	SBW feed blending																					
																					T-1	
37	Total volume and composite SBW feed composition																					
													K-1								T-1	
38	Noble metals concentration in SBW feed																					
																					T-1	
43	Composition and concentration of solids in SBW feed																					
																					T-1	
44	Required control of feed mix tank temperature																					
																					T-1	
45	Feed preparation requirements (excl. evaporation)																					
													M-6									
47	Melter offgas temperature																					
	A-22																					
48	Glass frit composition																					
													M-0									
49	Waste loading in glass																					
	A-2												M-0									
54	NOxidizer oxidation chamber chemistry																					
55	NOxidizer reduction chamber reaction and quench temperatures																					
																					Q-1	

D B E	CETL #1 Ref. 5	Hg/GAC tests Ref. 6	Melt rate tests Ref. 7	Rad crucible Ref. 8	CVS Ref. 9	Off-gas systems Ref. 10	RSM#1 Ref. 11	S Partitioning Ref. 12	Sulfate sep'n Ref. 13	MSE NOx tests Ref. 14	HILWE Ref. 15	VLE tests Ref. 16	RSM#2 Ref. 17	Rheology Ref. 18	Scrubber Ops Ref. 19	2nd Wastes Ref. 20	NOx model Ref. 21	Wet /dry feed Ref. 22	Off-gas FS Ref. 23	Feed char Ref. 24	WM-180 Ref. 25	Worst case Ref. 26
59										J-1							Q-1					
60						F-1				J-4												
66						F-5																
67						F-4							M-15									
68																P-4						
70		B-1																				
71		B-1																				
74		B-2																				
	A-9 A-17																					
75																						
	A-9 A-17																					
76																						
81																						
85		B-2																				
86																						V-6
95																						V-6
101																						
102	A-17																					
106																						
107																						
108																						

D B E	CETL #1 Ref. 5	Hg/GAC tests Ref. 6	Melt rate tests Ref. 7	Rad crucible Ref. 8	CVS Ref. 9	Off-gas systems Ref. 10	RSM#1 Ref. 11	S Partitioning Ref. 12	Sulfate sep'n Ref. 13	MSE NOx tests Ref. 14	HLLWE Ref. 15	VLE tests Ref. 16	RSM#2 Ref. 17	Rheology Ref. 18	Scrubber Ops Ref. 19	2nd Wastes Ref. 20	NOx model Ref. 21	Wet /dry feed Ref. 22	Off-gas FS Ref. 23	Feed char Ref. 24	WM-180 Ref. 25	Worst case Ref. 26
113	Target disposal sites for secondary wastes																					
																P-1 P-2 P-3 P-4						
118	Disposal requirements for mercury-containing secondary wastes																					
																P-1 P-3 P-4						
120	NOx abatement requirement in process offgas																					
																				S-1		
125	Disposal path for spent HEPA filters																					
																P-3						
126	Disposal path for GAC																					
																P-4						
133	Speciation of iodine in melter offgas																					
A-14																						
138	Safe handling of melter feed during abnormal shutdown																					
														N-4								
139	Selection of initial baseline process configuration																					
															O-1					S-1		
144	Melter feed rate basis																					
							G-1															
149	NOxidizer reduction chamber air requirement																					
																		Q-1		S-2		
152	Film cooler outlet temperature																					
A-19																						
155	Slagging/plugging control at melter outlet																					
A-25																						
157	Control of foaming in melter																					
A-26													M-3	N-0 N-1 N-2								
158	Speciation and partitioning of sulfate in melter																					
A-2 A-15							G-9 G-10 G-11	H-2					M-1 M-16									
159	Feed evaporation requirement																					
							G-1													R-1		
163	Fate of chloride and fluoride in the melter																					
A-10														M-12								

D B E	CETL #1 Ref. 5	Hg/GAC tests Ref. 6	Melt rate tests Ref. 7	Rad crucible Ref. 8	CVS Ref. 9	Off-gas systems Ref. 10	RSM#1 Ref. 11	S Partitioning Ref. 12	Sulfate sep'n Ref. 13	MSE NOx tests Ref. 14	HILWE Ref. 15	VLE tests Ref. 16	RSM#2 Ref. 17	Rheology Ref. 18	Scrubber Ops Ref. 19	2nd Wastes Ref. 20	NOx model Ref. 21	Wet/dry feed Ref. 22	Off-gas FS Ref. 23	Feed char Ref. 24	WM-180 Ref. 25	Worst case Ref. 26
	164	Should Sr-90 be extracted from scrubber blowdown																				
	167	Soluble species in offgas																				
174	A-18												M-17									
		NOxidizer reduction chamber chemistry using propane																				
						F-1				J-1 J-3								Q-1				

3. IMPACTS OF FY 2001 WORK TO DBES

This section provides a brief discussion of how the work performed in the last year affects Design Basis Elements (DBEs). Data that pertained to approximately 50 DBEs were obtained during the year. In many cases the new data substantiated earlier assumptions contained in the DBEs.

DBE-3 Detection of Sulfate Salt Layer in the Melter

Experience from FY 2001 crucible, research-, and pilot-scale melter tests suggests that formation of a segregated sulfate salt at some time during vitrification of SBW is likely to occur. Control of the growth and extent of the salt layer is necessary from several standpoints (safety, corrosion, removal and disposal of accumulated salt). Thus, unless the concentration of sulfate is reduced by preprocessing of the feed it will be necessary to monitor the formation and growth of any salt layer that forms on the surface of the glass in the melter. The method used to accomplish this in the pilot tests to date (manual probing of the melt with a refractory rod) is not viable for a production scale, remotely-operated melter. Therefore, a reliable, automated method for detecting the salt layer must be developed and demonstrated. A possible approach is to monitor electrical conductivity across the melt using a refractory probe. To be effective this approach would require a design that promotes preferential salt accumulation around the probe. This would ensure detection of the salt phase *before* development of a continuous layer across the surface of the melt and shorting of the electrodes.

DBE-16 Submicron Particle Size Distribution Out Film Cooler

Knowing the particle size distribution (PSD) of entrained solids in the melter offgas is necessary in defining requirements for design of equipment to remove the solids. Measurements of PSD made during the Clemson-1 test (Ref. 5) are the first measurements of solids in the offgas of melter tests of SBW surrogate waste. Based on SEM analysis of particulate samples from the Clemson-1 tests, three different particle shapes were identified, and particle sizes estimated for each, as well as for the aggregate. Additional data is needed to determine (a) the effects of feed and melter operating parameters on PSD of offgas solids, (b) the normal variation in PSD is during steady state operating conditions, and (c) the composition of the smaller particle fractions. It may also be necessary to characterize the behavior of "rod-like" particles with respect to removal in scrubbers, filters, etc. Such particles constituted roughly 25% of those collected in the Clemson-1 test.

DBE-17: Nitrate Destruction Chemistry in Melter

Nitrates and nitric acid in the waste are mostly destroyed in the melter by reaction with sugar, resulting in N_2 , N_2O , NO and NO_2 in the melter offgas. Small amounts of nitric acid can volatilize and nitrates can be entrained in the offgas. All melter tests in which offgases were analyzed for nitrogen species can contribute to this DBE. However, interpreting experimental melter offgas data to derive nitrate destruction chemistry is complicated by several factors. Off-gas samples are taken and hence reflect conditions not at the melter outlet but at points downstream, usually the film cooler outlet. Reactions continue to occur, most notably the shift of NO to NO_2 as the offgas is cooled from the melter outlet temperature or as the sample cools to room temperature, or oxidation and capture of NO on surfaces or in scrub solutions. Secondly, air in-leakage to the melter, which cannot be accurately measured, masks the amount of N_2 produced from nitrate destruction. And finally, as experienced in pilot testing during FY 2001, offgas analyses may not include all nitrate destruction products or may not be capable of measuring the low concentrations of some species.

Data from the Clemson-1 test (Ref. 5) includes measurements of NO , NO_2 , HNO_3 , N_2O and N_2O_4/N_2O_5 at the film cooler outlet (estimates of melter outlet composition were extrapolated from FC

dilution). The NO to NO₂ ratio seen in this test was 1:0.9. Data from this experiment could provide a basis for mass balance calculation of the amount of HNO₃ and N₂O in the melter offgas. The data shows that the concentration of N₂O₄ is small, and can either be neglected, or combined with NO₂ since N₂O₄ is an intermediate in the absorption of NO₂ by water.

Data from RSM-1 (Ref. 11) includes calculated NO, N₂O and total NO_x concentrations in the melter offgas based on analysis of samples downstream of the film cooler. However poisoning of the NO_x analyzer's catalytic reactor prevented obtaining NO₂ (or total NO_x) concentrations. The N₂O concentrations seen in RSM-1 were typically 24% of the NO concentration; an order of magnitude higher than the Clemson-1 data shows. The reason for this large difference in the NO_x chemistry is not known and may require additional testing.

NO₂ measurement difficulties experienced in RSM-1 were overcome for the RSM-2 test which provided measurements of NO, NO₂, and N₂O at the film cooler outlet (Ref. 17). The NO to NO₂ ratio with sugar as reductant in this test was 1:0.2, which differs by more than a factor of four from the Clemson-1 data. The NO:N₂O ratio for the two tests differed by a similar factor (1:0.017 vs 1:0.056 in Clemson-1 and RSM-2, respectively) but in the opposite direction. These results suggest significantly different nitrate destruction chemistry occurring in the Clemson and RSM melters.

DBE-18: Melter Chemistry

DBE-18 contains assumptions regarding the pyrolysis and oxidation of sugar in the melter and decomposition of carbonates that may be present in the waste. Clemson-1 offgas data (Ref. 5) shows nominal ratios of CO to CO₂ and H₂ to CO₂ in the melter offgas of 0.096:1 and 0.086:1. These data could be used to update DBE-18, which presently shows a CO to CO₂ ratio of 0.117:1 and a ratio of hydrogen to CO₂ of 0.013:1. However, the large variation in the measured CO₂ concentration (99%) makes this questionable, especially in light of the offgas data from RSM-1 (Ref. 11) which show average ratios of 0.21 for CO to CO₂ and 0.035 H₂ to CO₂, and from RSM-2 (Ref. 17) which show corresponding ratios of 0.10 and 0.027, respectively.

Data from Clemson-1, RSM-1, and RSM-2 could be used to provide a conservative basis for organics in the melter offgas, which are presently assumed to be zero. Clemson-1 data show an average 116 ppm THC in the melter offgas while the RSM-1 and RSM-2 data show 1140 and 700 ppm, respectively. The Clemson-1 data also includes analyses that identify many of the organic compounds present in the offgas.

DBE-19: Target Concentration of Total Dissolved Solids (TDS) in Melter Feed

The TDS concentration in the melter affects the melt rate and hence the melter size requirement. The higher the dissolved solids (or lower the water) in the waste, the higher the melt rate. However, high solids concentrations may cause problems in the feed system, such as precipitation, agglomeration and settling of solids in tanks and plugging in transfer lines. An engineering study was performed in FY-2001 (Ref. 22) that briefly discussed partial evaporation of SBW, but concluded that there was no clear benefit to partial evaporation. The study then did a more detailed evaluation of total evaporation (see DBE-159). Data from the RSM-1 (Ref. 11) test confirmed that the melt rate was limited by the water content of the feed.

Rheology studies (Ref. 18) showed that the present feed, without evaporation, has acceptable rheological properties. However, the rheology studies were performed with solids equivalent to a waste loading of 35% and not at lower waste loading. While questions remain regarding possible benefits from partial or complete evaporation of the melter feed, the data generated in FY-2001 support the present

assumption that the TDS of the melter feed, assuming no evaporation, would be acceptable for a 35% waste loading. In RSM-2 tests, however, which were run at 20% waste loading using formulation “SBW-22”, addition of GFCs to the SBW simulant resulted in gelling and required addition of either nitric acid or water to enable transport of the feed via the RSM-2 feed system.

The target concentration of TDS in the feed will depend upon whether GFCs or glass frit is used. The evaluation between frit and GFCs has yet to be performed and documented.

DBE-20: Reductant Selection and Concentration in Melter Feed

In work performed in the past year, criteria were defined for selection of a reductant (Ref. 7), followed by laboratory-scale slurry-fed melter tests at PNNL and melt rate furnace tests at CETL. These tests evaluated four reductants – starch, sugar, carbon and glycolic acid. The test results showed that only sugar met the reductant performance criteria (entry C-5 in Table 2-1). RSM-2 (Ref. 17) tests suggest that glycolic acid is unsuitable as the sole reductant in the feed. However, the results may also suggest that use of sugar and glycolic acid *together* may be superior to either alone (based on the observation that glycolic acid is more effective as a gas phase reductant while sugar is more effective in controlling redox in the glass (see entries M-2 and M-9 of Table 2-1).

The concentration of reductant in the melter feed affects several performance parameters, namely glass properties, foaming in the melter, glass melt rate, nitrate reduction, sulfate content of the glass and whether a salt layer forms and grows on the melt. While the test data shed some light on the relationships between reductant concentration and these performance parameters, it is unclear regarding others. For example, crucible tests indicated that the sugar concentration (varied from 77 to 164 g/L of SBW at a 35 wt % loading and SBW-2 frit) had little effect on salt formation (entry E-5 in Table 2-1). However, the data from RSM-1 tests showed increased reductant levels mitigated molten salt accumulation at similar waste loadings (entry G-11 in Table 2-1). RSM-2 tests showed that no salt layer was observed with the SBW-22 formulation with no sugar at all in the feed (entry M-1 in Table 2-1). The redox ratio at which metal oxides in the glass are reduced and form a separate molten or crystalline phase sets an *upper* limit for reductant concentration. There may also be a point at which glass durability is unacceptably compromised by too high a reductant concentration (and redox ratio). However, presumably there are *lower* limits on redox ratio (and corresponding reductant concentration in the feed) required to control foaming, melt rate, and sulfate layer formation and growth to within acceptable ranges.

Data from tests completed to date should be evaluated *as a whole* to (a) determine whether credible correlations can be identified between feed sugar concentration and the above performance measures, (b) decide what additional testing may be needed, and (c) understand the chemistry and attempt to formulate a credible model that might be tested against the data. Once valid correlations are in hand they can be used to optimize the reductant concentration.

DBE-22: Pumping Behavior of Melter Feed Slurry

Initial rheology studies (Ref. 18) determined viscosities and settling and mixing characteristics for simulated melter feeds made with four types of silica plus glass forming chemicals added to liquid SBW simulant. The testing confirmed that (a) addition of sugar, GFCs and particulate silica to the feed increased the "effective" viscosity (by a factor of nearly 2), and (b) the finer the solids the longer the settling time and the easier it would be to maintain a solid frit in suspension. Testing also indicated that addition of iron oxide may inhibit reactivity of the blended feed and that feed with solid frit will be erosive.

Additional testing with actual frit for SBW-22 (or other formulations determined appropriate for the different SBW waste compositions) should be done using an actual mixing and feed system to determine parameters that ensure a reliable, homogeneous feed to the melter. The above observations and data could be used in scoping the testing program. Test data would be used to provide impeller size, speed, and position and minimum tank level required to provide a homogeneous feed. Additional information that is needed include erosion effects of feed on the expected materials of construction, the changes to feed rheology that occur over time as reactions proceed (particularly precipitation reactions), and a determination of whether frit should be used in place of GFCs.

DBE-24 Partition Factors for Feed Species in Melter

DBE-24 contains partition factors assumed in the mass balance for all nonvolatile and semivolatile species in the melter. Species entrained in the melter offgas may be particulate in the feed entrained in the offgas, solids crystallized from the feed and entrained in the offgas, molten glass splattered into the melter vapor space and carried into the offgas, or volatile species vaporized in the melter and then condensed as the offgas cools.

Tests at Clemson (entries A-4 through A-8 in Table 2-1) and Hanford (entries G-5 through G-8 and M-10 through M-14) contain melter DF factors, the reciprocal of partition factors. DF factors calculated from melter test data can contain considerable uncertainty because of the lack of closure in the mass balance. For Clemson-1 tests, the melter DF factors derived from glass analyses and feed measurements and reported in the Ref. 5 differ greatly (for some species by more than an order of magnitude) from those calculated on the basis of the quantities of individual species actually measured in the output from the melter (i.e., glass, scrub solution, and offgas solids). If the mass balance were completely closed then the two sets of DFs would agree. The fact that there are large discrepancies may be due to the lack of mass balance closure, analytical error, or measurement error and serves to underscore the large uncertainties in the reported DFs and the need for better measurements and (possibly) better mass balance closure.

From data collected in the RSM tests, "aerosol" and "total" DFs were calculated. Aerosol DFs were calculated by analysis of solids collected on filters only, while total DFs were calculated using analyses of scrub solutions and filtered solids data. Because equipment scale and configuration affect DF factors, many of the DF factors derived from RSM or even Clemson data will be different from those of a full-scale melter. For example, the RSM-1 and -2 data for Cs show a total DF of 3-5 and 9-49, respectively, and the Clemson data show a DF of 2, but data from the DWPF melter show a Cs DF of about 50 (Ref. 32).

Additional data are required to improve mass balance closure and reduce the ranges of between the upper and lower bounds for melter DFs, particularly for species that determine offgas treatment design requirements (e.g., Cs, Tc, Sr, Cl, F, TRU radionuclides). Data are also needed to extrapolate the effects of operating parameters and melter scale on melter DFs.

DBE-25 Iodine-129 Concentration in SBW Feed

The updated SBW composition (Ref. 24) shows an average ^{129}I concentration of $5.5\text{E-}8$ Ci/liter, and ^{129}I concentrations in different tanks of $1.3\text{E-}8$ to $9.9\text{E-}8$ Ci/liter. Historical analytical data for the Tank Farm was reviewed in April, 2001 by Mike Swenson, who found ^{129}I concentrations of $1.1\text{E-}8$ to $6.6\text{E-}8$ Ci/liter (see DBE titled "Iodine-129 concentration in SBW feed" in Appendix C of Ref. 28). The concentrations shown in Ref. 24 are thus consistent with the historical data.

DBE-27 Partitioning of Acid Gases and Hg During Quench/Scrubbing Operations

Two tests provided additional understanding of partitioning of mercury during scrubbing that has significant implications for the treatment flowsheet. Mercury VLE data from the first of these tests (see entry L-1 of Table 2-1) indicates that the quantity of oxidized mercury (HgCl_2) that will partition to the gas phase is 10-100 times the amount predicted from thermodynamic considerations. However, even with this unexplainably high mole fraction of mercury in the gas phase, the portion partitioning to the liquid is much higher than was assumed in theoretical mass balance calculations. Based on this data alone, the mercury (if fully oxidized) should build up to much higher concentrations in the scrub than was indicated in the mass balance.

In the RSM-2 test (entry M-15 of Table 2-1), no mercury was found in the offgas downstream of the HEME. The CEM data from this test show that a significant portion (97%) of the mercury detected in the offgas directly downstream of the melter was captured by the scrub system upstream of the HEME exit. Below 485°C thermodynamic calculations indicate that HgCl_2 is the favored form for mercury if Cl is present in excess. However, the RSM-2 data clearly indicate elemental Hg as the dominant form. This may be because Hg oxidation kinetics become limiting as the temperature falls, preventing the predicted shift from elemental Hg to HgCl_2 . This possibility is consistent with results of other studies (see Ref. 27). Based on the preponderance of the elemental form of Hg one would expect poor mercury scrubbing efficiency. However, the RSM-2 data suggest that scrubbing efficiencies of 73-97% were achieved. A mechanism to explain the high apparent scrubbing efficiency is needed.

For the overall RSM-2 demonstration, 27% of the Hg fed to the melter was found in scrub solutions and 46% was found in acid flushes of the HEME. From this it may be conjectured that at least 63% of the mercury scrubbed from the offgas was elemental and insoluble (consistent with the speciation measurements). One explanation of the high scrubbing efficiency is that the combination of oxidizing conditions (high NO_2 and chloride in the gas phase combined with a HNO_3 scrub solution) together with effective liquid/gas contact in the HEME were sufficient to oxidize most of the Hg, which was then dissolved (presumably as HgCl_2) in the liquid on the HEME mesh. A second possibility, based on the apparently high fraction of insoluble Hg scrubbed, is that elemental Hg in the offgas condensed in the HEME and then formed particulate matter (through homogeneous or heterogeneous nucleation) which was collectable through intimate contact between the gas and liquid in the HEME mesh.

If the high fraction of detected mercury in the scrub liquids is representative of the mercury fraction that will be scrubbed in a production scale system, then alternative scrub processing methods should be evaluated. Otherwise, unless mercury is removed from the scrub, the grouted scrub purge may contain an excessive amount of leachable mercury. In addition, the volume of GAC required for Hg polishing may be lower than previously expected.

DBE-29 Carryover of Melter Feed to Off-gas

The carryover of melter feed to the offgas describes the fraction of the total input stream that is entrained in the offgas leaving the melter before it undergoes significant reaction and/or volatilization. Thus, it describes (a) the mass fraction of the non-volatile species in the feed which partitions to the offgas, and (b) the mass fraction of volatile and semivolatile species which partitions to the offgas *in addition to* that which would volatilize without any feed carryover. The carryover value therefore provides the default lower limit to the fraction of each species in the feed that is expected in the melter offgas.

Both the RSM and the Clemson-1 tests provided DF information for the melter. Entry A-4 in Table 2-1 gives a range of 10-28 for the overall DF for Clemson-1 and entries G-7 and M-13 give ranges of 1100-1230 and 58-1560 for the DF of non-volatile species. Both should be equivalent to the carryover of melter feed. Due to the large difference in these ranges we cannot confidently choose one or the other

(or even an average) as "representative" of a production scale melter. The best we can do at this time is to use 10 as a lower bound and 1560 as an upper bound on the overall DF for carryover mass. Unfortunately this leads to a large range on the possible quantities of secondary waste that must be treated and disposed. Additional testing is needed to determine (a) whether melter design factors or poor data quality are responsible for the large disparity in the above data, and (b) better upper and lower bounds on the expected carryover range for a full-scale system.

DBE-30 Representativeness of Cold Simulants in Melter Tests

The first effort to compare glasses produced from simulants and from actual waste was made in FY 2001 (see Ref. 8). Radiologically cold and hot glasses were prepared in laboratory crucible melts using the WM-180 surrogate with two frit formulations, and also with actual WM-180 waste and the same two frit formulations. The compositions of the resultant cold and hot glasses were analyzed and found to be comparable. Leaching properties of the radioactive glasses were measured and were found superior to those of EA glass. XRD measurements of the radioactive glasses gave no evidence of crystallization. Similar measurements have been done on cold glasses in other testing programs (e.g., see entries A-11 and G-3 in Table 2-1) with similar conclusions, though this was not done here.

Thus, the data produced from this effort support the premise that the simulants are representative of real SBW from WM-180. However, additional data are needed to confirm this preliminary conclusion. Specifically, offgas measurements should be made with comparable setups for cold and hot feeds and the results should be compared. In addition it would be desirable to compare the behavior of noble metals and semivolatile radionuclides or their non-radioactive surrogates (e.g., non-radioactive Cs, and Rh for ⁹⁹Tc) in cold and hot glasses. Finally, better estimates of the composition of UDS present in SBW should be made and simulated in non-rad testing.

DBE-31 Composition Envelope for Acceptable Glass

Progress has been made in defining glass compositions that adequately immobilize SBW. Building on data gathered in previous years that relate glass properties to glass composition, a series of ten GFC formulations were defined and tested for WM-180 simulant (see entry E-0 in Table 2-1), leading to the recommendation to use the SBW-9 formulation in the RSM-1 and Clemson-1 tests. Data from these tests confirmed that the glass properties using this formulation were acceptable. In response to an update in the WM-180 composition, additional frit formulations were developed and tested that resulted in the recommendation of the SBW-22 formulation to be used in the RSM-2 test. Additional compositional variation studies will be required in order to define glass formulations that fully envelope the SBW composition range.

DBE-34 Frit vs GFC in Melter Feed

No tests using a glass frit have yet been performed. However, an evaluation of different forms of silica that could be used with other GFCs was done this past year (see entry N-0 in Table 2-1 or Ref. 18), and provides melter feed rheology data for the melter feed with GFCs. Similar tests (in addition to melter tests) are now needed using frit in place of GFCs.

DBE-35 SBW Feed Blending

The updated SBW composition report (Ref. 24) provides compositions of SBW in existing tanks as well as blended in future (new) tanks. Blending the SBW would produce an estimated 68 m³ less glass (through blending down of sulfur concentrations) and 27 m³ less grouted waste (see Attachment 3). In addition, a single, blended waste stream would require qualification of fewer glass formulations or the

qualification of a narrower composition envelope. From these standpoints blending of SBW tank-by-tank processing would be desirable. However, the method(s) by which such a blending scheme could be accomplished have not been scrutinized and no programmatic decision or guidance has been provided.

DBE-37 Total Volume and Composite SBW Feed Composition

The updated SBW composition report (see entries T-0, T-1 in Table 2-1 or Ref. 24) provides the most recent estimates of the total volume and compositions of SBW that would be fed to the melter. These estimates are projections to specific future proposed startup dates for the vitrification facility, and as such, depend on INTEC Tank Farm management plans and projected generation rates of new wastes, as well as the vitrification startup date. While changes in Tank Farm management plans and waste generation operations will affect the composition of individual tanks and will be used to update the projected volumes and compositions, the effect on the average or blended composition is expected to be small. It is intended that the SBW composition report be periodically updated as wastes in the different tanks are evaporated, blended or new waste added, and new waste samples (both liquid and solids) are obtained and analyzed.

DBE-38 Noble Metals Concentration in SBW feed

The concentration of noble metals in the feed is of interest because of the possibility of their forming a separate molten phase on the bottom of the melt under reducing conditions. This, in turn, would likely lead to melter failure by shorting the electrodes. Noble metal concentrations in SBW are given in Ref. 24.

DBE-44 Required control of feed mix tank temperature

Once sugar is added to the SBW feed, temperature control is required to avoid initiating the reaction between sugar and nitric acid. One FY 2001 test showed that no observable reaction occurs below 50°C (see entry N-4 in Table 2-1). On this basis is unlikely that any temperature control would be needed in the mix tank unless heating is required for blending or dissolution, or unless heat must be added or removed due to reactions between the waste and GFCs. We assume that a minimum 10°C safety margin will be maintained so that the feed tank temperature should be controlled to <40°C. The adequacy of this temperature margin will need additional verification under all expected feed compositions under carefully monitored test conditions.

DBE-45 Feed Preparation Requirements (Excluding Evaporation)

Feed preparation requirements include mixing requirements for homogenization of SBW liquid plus additives (frit plus reductant and [possibly] nitric acid—see entry M-6 in Table 2-1). It also includes the form and properties of the additives prior to mixing. The DBE presently assumes sugar will be added as a syrup (67 wt% sugar solution), however tests to date have used granular sugar, and an analysis of process alternatives (see Attachment 3) recommends the use of granular sugar. Assuming use of granular sugar, testing will be needed to establish the mixing requirements to effect complete dissolution and blending. Acceptable forms of silica relative to mixing and settling requirements have been identified (Ref. 18). Testing may be required to establish that settling of frit does not occur in quiescent regions of the mix tank.

In a separate study (Ref. 13), removal of sulfate from the feed was demonstrated. Since the glass waste loading is limited by the sulfate concentration in the waste (Ref. 12), removal of sulfate would permit an increase in glass waste loading. Costs associated with additional unit operations and treatment and disposal of the radioactive BaSO₄ waste stream need to be compared with the costs of making,

storing, and disposing more glass. Additional discussion of sulfate removal from the feed is given in the above-referenced analysis of process alternatives.

DBE-47 Melter Off-gas Temperature

Although not a primary objective of the test, conditions were varied during the Clemson-1 melter test (Ref. 5) to achieve both “low” and “high” plenum temperatures (entry A-22 of Table 2-1). The target plenum temperature maintained throughout most of the test was 600°C; the temperature was reduced to about 450°C for the low temperature tests and raised to 650°C for the high temperature test. Reducing the plenum temperature reduced the NO_x concentration in the melter offgas by about 20% and the SO₂ concentration by about 65%. Lower plenum temperature means less air and steam are required in the film cooler resulting in reduced offgas flow. This reduces the offgas system size requirement. However, additional data from low plenum temperature melter tests is needed to (a) corroborate the above observations from Clemson-1, and (b) quantify “penalties” that could be associated with reduced plenum temperature (e.g., reduced melt rate, increased salt layer on melt, increased H₂, CO, and hydrocarbons in the offgas). Based on the experience of DWPF, the lower limit is set by flammability concerns, i.e., a high enough temperature is needed to ignite combustible gases in the melter to avoid flammability concerns in offgas equipment. This, together with other the issues identified through additional testing, would have to be considered.

DBE-48 Glass Frit Composition

Ten frit formulations were tested in the glass compositional variation studies (CVS) (entry E-0 of Table 2-1), using waste loadings of 25-45%. Based on the results of this study, a waste loading of 30 wt% and a frit formulation, labeled “SBW-9”, of 65 wt % SiO₂, 15 wt % B₂O₃, 5 wt % Li₂O, 10 wt % Fe₂O₃, and 5 wt % CaO was recommended for testing in the RSM-1 and Clemson-1 tests using an early simulant for WM-180 SBW. Subsequent to this study, a revised simulant composition for WM-180 waste was developed, based on new sulfate measurements for the tank liquid (Ref. 25), that had a sulfate content 37% higher than had been used in the earlier simulant. Additional frit formulations were tested with the goal of retaining a higher amount of sulfate in the glass (Ref. 9) and minimizing the growth of a sulfate salt layer in the melter. The SBW-22 formulation that resulted from these efforts had the following composition: 67.95 wt % SiO₂, 6.03 wt % B₂O₃, 6.11 wt % Li₂O, 1.52 wt % Fe₂O₃, 5.02 wt % CaO, 1.75 wt % MgO, 4.29 wt % Na₂O, 4.88 % V₂O₅, and 2.44 wt % ZrO₂. This formulation was used in the RSM-2 test (Ref. 17). No sulfate salt layer was observed in the melter during RSM-2 testing with the new formulation at a sugar concentration of 160 g/L (entry M-1 in Table 2-1). On this basis the “preferred” formulation (as of this writing) is presumed to be SBW-22. However, glass composition optimization is not yet complete and will require completion of the composition variation study and evaluation of the range of waste compositions likely to be processed in the melter.

DBE-49 Waste Loading in Glass

Glass formulation studies (entry E-0 of Table 2-1) performed in 2001 included melts with waste loadings from 25-45% and based on glass property results, recommended a waste loading of 30% for WM-180, based on the objective to minimize sulfate salt layer formation in the melter. One objective of the RSM-1 tests (Ref. 11) was to determine if the waste loading might be increased, and glasses were prepared using waste loadings of 30, 32 and 35 wt%. Later on, new analyses of WM-180 waste showed a higher sulfur content than had previously been measured. Prior work had indicated that the sulfate content of the waste would limit the glass waste loading if an assumed requirement of the glass formulation is that no salt layer form in the melter (entry H-2 in Table 2-1). Thus, subsequent tests performed with the higher sulfate WM-180 surrogate (Ref. 25) used lower waste loadings. RSM-2 tests

confirmed that salt layer did not form (entries M-1 and M-16 in Table 2-1) using waste simulant with 0.07 M sulfate and a waste loading of 20%.

The results of studies performed during and before FY2001 suggest that the loading of SBW in glass is primarily determined by the allowable sulfur concentration. Current estimates are that between 0.8 and 1.0% of sulfur (on the basis of wt% of SO₃ targeted in the glass) would be the acceptable limit assuming most sulfur in the offgas is recycled back to the melter and that the accumulation of a molten salt layer in the melter is unacceptable. This translates into a loading of roughly 20 wt% for the current WM-180 waste composition. Since the total quantity of glass produced is inversely proportional to waste loading, reduction from 30-35% to 20% waste loading has significant negative impacts on the SBW vitrification project cost and schedule. Additional discussion of schemes utilizing increased waste loadings is given in Attachment 3.

DBE-54 NOxidizer Oxidation Chamber Chemistry

NOxidizer tests at MSE (entries F-1 and J-2 in Table 2-1) demonstrated >99.7% destruction efficiency of benzene-spiked feed. The stack THC analyzer for these tests registered below its detection limit. Kinetic modeling (Q-1 in Table 2-1) of the NOxidizer showed outlet CO and H₂ concentrations equivalent to 99.9996% and 99.999% removal respectively.

DBE-55 NOxidizer Reduction Chamber Reaction and Quench Temperatures

Preliminary modeling of NOxidizer performance by the equipment vendor (entry S-2 of Table 2-1) shows a reduction chamber temperature of 2400°F (1316°C) and a quench temperature of 1665°F (907°C). Kinetic modeling (Row Q-1 of Table 2-1) shows a reduction temperature of 2314°F (1268°C), and a quench temperature of 1672°F (911°C). The kinetic modeling was based on natural gas rather than propane fuel. DBE-55 currently specifies a reduction temperature of 1200°C and a quench temperature of 820°C.

In NOxidizer tests at MSE (Ref. 14) the quench temperature was one of three process variables studies relative to optimizing NO_x destruction efficiency. The highest NO_x destruction efficiencies were seen with (a) low quench temperature (1600°F), low temperature difference between the quench and reoxidation stages (50°F) and high oxygen in the NOxidizer effluent (>4%), (b) low quench temperature (1600°F), high temperature difference (200°F) and low oxygen (1.5%), and high quench temperature (1800°F), low temperature difference (50°F) and low oxygen (1.5%). Thus it appears that high NO_x destruction efficiencies can be achieved over a wide range of quench temperatures (1600-1800°F) by adjusting other variables.

Based on additional information from the NOxidizer vendor, an air rate to the reduction chamber much greater than currently assumed is required to maintain burner stability (entry S-2 in Table 2-1). With a higher air rate, a higher fuel rate will be required to maintain reducing conditions, and the reduction chamber temperature will increase from that shown in the current mass balance (Ref. 28). It should be noted that quench water is added both in the reduction chamber and the quench chamber, and the “reduction chamber temperature” includes the cooling achieved by reduction chamber quench water.

DBE-59 Speciation of Mercury in NOxidizer Effluent

NOxidizer tests at MSE (entry J-4 of Table 2-1) confirmed the present assumption that mercury is reduced in the NOxidizer to elemental mercury. The tests showed greater than 99.8% of the mercury in the NOxidizer effluent as elemental.

DBE-60 pH Control in Scrubber Solutions

In the RSM-2 tests (entry F-5 in Table 2-1) the acid concentration of the initial scrubbing solution was adjusted to around 1. M. Over the course of the tests the concentration gradually decreased to ~0.6 M suggesting that alkaline solids carried over from the melter to the scrub neutralized acids (such as HNO_3 and HCl) captured or formed in the scrub by absorption of acid gases as well as part of the initial acid in the scrub. Thus it is likely that a controlled addition of acid to the scrub will be required. The optimal pH, however, has not been determined. If the acidity is high then most of the solids collected may likely dissolve, precluding a selective recycle of solids only. Selective recycle of solids may be desirable, given the apparent solubility of sulfate (see entry G-10 in Table 2-1).

DBE-64 Air Addition to NOxidizer Oxidation Chamber

The offgas treatment feasibility study (Ref. 23) includes a price quotation from the NOxidizer vendor along with the vendor's estimates of equipment requirements and flow rates. The vendor's estimate of airflow to the oxidation chamber confirms the present assumptions of the mass balance.

DBE-66 Mercury Speciation Downstream of Melter

(See discussion of DBE-27 Partitioning of Acid Gases and Hy During Quench/Scrubbing Operations)

DBE-67 Peak Acceptable Mercury Loading in GAC

Test data from Nucon (entry S-3 of Table 2-1), the supplier of Mersorb, shows a mercury loading of 23 gm Hg per gm carbon in the first section of a bed divided into 6 sections. The loading drops to 14 gm/gm in the fifth section of 0.3 gm/gm in the final section. The Nucon tests were performed using an air stream. These loadings should be verified by testing a gas stream which simulates vitrification offgas, in conjunction with TCLP testing of the loaded carbon to determine whether it can be disposed without further treatment.

DBE-68 Residence Time Requirement for Hg Extraction in GAC

Test data from Nucon (entry S-3 of Table 2-1), the supplier of Mersorb, show a residence time of 10 seconds is required to avoid mercury breakthrough over the time required to process SBW (400 operating days). The Nucon tests were performed with an air stream. Small scale mercury removal tests using simulated vitrification offgas (entry B-0 of Table 2-1) were performed at residence times of 0.1 to 0.53 seconds and achieved 99% removal of the Hg in the gas (entries B-0, B-1 in Table 2-1).

DBE-70 Mercury Removal Efficiency of GAC Beds

In five small-scale GAC bed tests (entries B-0, B-1 in Table 2-1), removal efficiencies varied between 99.1% and 99.6%. Gas composition, residence time, and speciation of mercury were varied in these tests; however, the effects of these variables on mercury removal is not clearly evident from the resulting data. In addition, while the effluent concentrations from the tests were within MACT emissions limits, the inlet concentrations were approximately 100 times less than those indicated by the current mass balance. No data have been generated indicating how removal efficiency varies with inlet concentration. Vendor data based on removal of mercury from air (contained in Ref. 23) indicate that the removal of mercury is 100% until breakthrough, and the time to breakthrough varies with residence time. Additional testing and/or modeling is thus needed to clearly determine the effects of inlet mercury speciation,

concentration, and residence time on mercury removal efficiency and time (or volume throughput) to breakthrough.

The effect of inlet Hg concentration on removal efficiency is also related to the question of scrubber efficiency in extracting Hg from the offgas. Mercury was not detected downstream of the HEME in the RSM-2 tests (entry M-15 in Table 2-1). If these data are representative then the high Hg scrubber inlet concentrations reported in the mass balance (attributable to low assumed Hg scrubbing efficiency) are erroneous and in need of adjustment.

DBE-71 Maximum Allowable NO_x Concentration in GAC Bed Influent

The feed gas in small-scale GAC tests (entry B-0 in Table 2-1) included 990 ppmv NO, 97 ppmv NO₂ and 1400 ppmv N₂O. Compared to the SBW vitrification process mass balance, the NO concentration in the test gas was 80% higher and the NO₂ concentration several orders of magnitude higher. Hence it appears the upper limit of NO_x in the GAC test bed influent is significantly above vitrification offgas normal levels. At this level the NO_x concentrations used in the test did not appear to affect mercury removal efficiency over the duration of the tests (14-26 hours).

However, based on NO_x concentrations in melter offgas determined from pilot testing (see entries A-14, G-13 in Table 2-1) potential concentrations of NO_x in the offgas entering the GAC beds during NOxidizer upset conditions could be significantly higher than what was tested. Additional testing may thus be appropriate to determine the behavior of the GAC bed at NO_x concentrations representative of such upset conditions (possibly as high as 2.4%).

DBE-74 Solids Scrubbing Efficiency in Acid Quench

(See discussion of DBE-75 Solids Scrubbing Efficiency in Acid Venturi Scrubber)

DBE-75 Solids Scrubbing Efficiency in Acid Venturi Scrubber

Based on entry A-9 in Table 2-1 the nominal measured scrubber efficiency in the Clemson-1 test was 0-50%. This number seems low, based on the authors' experience, but not inconceivable, depending on the size of the solids being collected. Hence, the more meaningful measurement from this test is probably the PSD in entry A-17. The efficiencies of industrial scrubbers are usually fairly well characterized as functions of the PSD. Rather than rely on the Clemson-1 measurement (which has a high uncertainty due to 83% closure on the mass balance (entry A-23 in Table 2-1) a more credible approach would be to corroborate the PSD measurements through additional testing and secure representative vendor performance data for candidate scrubber types.

DBE-76 Solids Scrubbing Efficiency in Caustic Quench

(See discussion of DBE-75 Solids Scrubbing Efficiency in Acid Venturi Scrubber)

DBE-81 Effects of 'Other' Species on Hg Polishing Effectiveness of GAC

Small-scale tests were performed (entry B-0 in Table 2-1) using a feed gas of wet air and a feed gas representative of vitrification offgas, containing CO, CO₂, NO_x, SO₂ and HCl in addition to N₂, O₂ and H₂O. Results indicate no detrimental effect from these other species (entry B-2 in Table 2-1).

DBE-85 Maximum Allowable Cl, F Content of Scrubbing Solutions

(See discussion of DBE-86 Fluoride Concentration Requiring Use of Additives for Corrosion Control)

DBE-86 Fluoride Concentration Requiring Use of Additives for Corrosion Control

In association with Jerry Christian's work on developing simulants for a "worst case" SBW simulant he examined the issue of fluoride concentrations that would raise corrosion concerns (entry V-6 in Table 2-1). From corrosion tests for the Fluorinel Dissolution Process on 304L stainless steel and past determinations of fluoride complexation constants (Refs. 30, 31) it was determined that a concentration of 0.055 M free HF at 35°C would limit the long term corrosion rate to 2 mils/yr and the weld corrosion rate to an acceptably level. Using this information together with the composition information for "worst case" SBW (documented in Ref. 26), an expected "worst case" concentration of free HF in SBW is about 5.65×10^{-4} M, which is well below the above the critical value of 0.055 M. Thus, it is safe to assume that there is no need for additives (such as $\text{Al}(\text{NO}_3)_3$ or H_3BO_3) in SBW feed to the melter process.

For downstream scrubber operations the data supplied by Dr. Christian may also be used to determine whether the same conclusion applies to scrub solutions. Based on mass balance calculations, the *total* concentration of fluoride in scrub liquors can be compared with the 0.055 M threshold. Based on apparently high retention fractions of fluoride in the glass (entry A-10 in Table 2-1) it appears unlikely that scrub fluoride concentrations would reach this level. If they do, it would then be necessary to repeat the fluoride speciation calculations done for the SBW feed using the scrub liquor composition to determine whether the free HF concentration is below the threshold limit.

DBE-101 Settling Rate of Solids in Scrubber Blowdown Liquid

No explicit measurements of scrub solids settling rates were made. In principle these rates could be calculated from PSD data (entry A-17 in Table 2-1), assuming all solid particles are spherical and non-agglomerating. However, comments regarding observed particle morphology make questionable the accuracy of such calculations. In addition, analysis of scrub solutions suggest that a significant fraction of the solids are soluble, which further confounds the problem of calculating a settling rate. In summary, there is little data from the FY 2001 that sheds meaningful light on this DBE.

DBE-102 Will Facility Comply with MACT?

Requirements for offgas treatment are discussed in the Off-gas Feasibility Study (Ref. 23), and include MACT. MACT is a joint rule-making effort under the Clean Air Act and the Resource Conservation and Recovery Act to establish emission standards for hazardous waste combustor sources. The MACT standards reflect the maximum degree of reduction in emissions of hazardous air pollutants that is achievable, taking into account control costs, non-air related health and environmental impacts, and added energy requirements. One significant impact of MACT is the "derived" requirement for NO_x removal. While NO_x is not regulated under MACT, NO_x removal is probably required in order to measure VOCs and SVOCs and thus demonstrate that *they* are within MACT limits.

Implications of offgas measurements from RSM-1, EV-16, and RSM-2 tests are discussed in Ref. 27.

DBE-106 Waste Acceptance Criteria For Mixed Waste Disposal at NTS

Major waste acceptance criteria were captured and summarized for the NTS disposal site. These criteria deal with listed waste codes, RCRA metal content, free liquids, compressive strength requirement for solid wastes, container requirements, and specific radionuclide concentration limits (see entry P-7 in Table 2-1).

DBE-107 Waste Acceptance Criteria For Mixed Waste Disposal at Envirocare

Major waste acceptance criteria were captured and summarized for Envirocare. These criteria deal with listed waste codes, RCRA metal content, free liquids, compressive strength requirement for solid wastes, container requirements, and specific radionuclide concentration limits (see entry P-5 in Table 2-1).

DBE-108 Waste Acceptance Criteria For Mixed Waste Disposal at Hanford

Major waste acceptance criteria were captured and summarized for the Hanford disposal site. These criteria deal with listed waste codes, RCRA metal content, free liquids, compressive strength requirement for solid wastes, container requirements, and specific radionuclide concentration limits (see entry P-6 in Table 2-1).

DBE-113 Target Disposal Sites for Secondary Wastes

Herbst, Kirkham, and Losinski performed an evaluation of disposal sites for each of the secondary wastes (entries P-0 to P-4 in Table 2-1). The grouted scrub purge can be disposed of at Hanford as either a Category 3 waste, or if ⁹⁰Sr is removed, Category 1 waste. Because the mercury concentration in the grouted scrub is greater than 260 mg/kg, a demonstration of treatment equivalency (to BDAT) will be needed to dispose this waste.

The vitrification facility has two sets of process offgas HEPA filters plus other heating, ventilation and air conditioning (HVAC) air filters. The evaluation recommends using washable filters to avoid having to dispose of highly contaminated upstream offgas HEPA filters. Otherwise these filters would require treatment (e.g., in the existing NWCF filter leach facility) prior to disposal. The downstream offgas HEPA filters are expected to be directly disposable at Envirocare. Additional information is needed to evaluate disposal of outlet HVAC filters.

Spent activated carbon could be disposed at Hanford, the Nevada Test Site or Envirocare. As with the grout, a treatment equivalency demonstration will be needed that is acceptable to the disposal site and regulators. The report recommends vitrifying spent ion exchange sorbent rather than dispose of it as a secondary waste.

DBE-118 Disposal Requirements for Mercury-Containing Secondary Wastes

To dispose of secondary wastes containing mercury two requirements (see entries P-1, P-3, and P-4 in Table 2-1) must be met: (a) The waste must pass the TCLP test for mercury, and (b) A demonstration of equivalency agreement with each disposal site must be negotiated. A demonstration of equivalency establishes that a treated RCRA waste will perform at least as well as the BDAT for that waste.

DBE-120 NOx Abatement Requirement in Process Off-gas

NOx abatement in the melter facility offgas will be required to a level negotiated with the State of Idaho, Department of Environmental Quality, Air Permitting Division (entry S-1 in Table 2-1).

Reference 23 discusses specific requirements for NO_x abatement in order to: (a) permit accurate analysis of VOCs and SVOCs, (b) avoid interference with Hg removal by activated carbon, (c) not exceed the State of Idaho significant NO_x emission level, (d) comply with the precedent for removal of NO_x set at other DOE high level waste facilities, i.e., the West Valley Demonstration Project, and (e) reduce plume visibility. The Feasibility Study concludes that if either a NO_x oxidizer or SCR reactor is included in the flowsheet, all requirements for NO_x abatement will be met.

DBE-125 Disposal Path for Spent HEPA Filters

(See discussion of DBE-113 Target Disposal Sites for Secondary Wastes)

DBE-126 Disposal Path for GAC

(See discussion of DBE-113 Target Disposal Sites for Secondary Wastes)

DBE-133 Speciation of Iodine in Melter Off-gas

Concentrations of HI, I₂, and total I were reported for one pilot scale test (Clemson-1, see entry A-14 in Table 2-1). However, it has been noted^d that the EPA Method 0050 which was used has not been validated as being able to separate HI and I₂. In addition, it will be noted from the reference entry in Table 2-1 that the HI and I₂ concentrations (0.096 and 1.26 ppm) sum to only about half the total I concentration (2.6 ppm), raising questions about the accuracy of the measurements. Finally, Ref. 5 states the question of iodine speciation was not addressed in the test because a suitable analytical method to determine partition factors for iodine was lacking. Nonetheless it would be informative to compare the measured quantity of iodine in the offgas and glass (from the reported total I concentration) with the quantity input to the melter in the feed. (Ref. 5 states that the iodine in the feed was increased from the nominal level for SBW-9 simulant. However, the actual concentration was not reported.) This comparison might indicate the fraction of iodine that can be expected to be retained in the glass, the fraction that partitions to offgas solids, and the fraction that remains in the gas phase. Determining both the iodine concentration in the offgas and its speciation is crucial to ensuring that secondary wastes meet waste disposal criteria.

DBE-138 Safe Handling of Melter Feed During Abnormal Shutdown

To detect reactions between sucrose and nitric acid, simulated SBW was mixed with sugar, stirred, purged with argon and the purge gas analyzed (entry N-4 in Table 2-1). The experiment was run at ambient temperature for several days, and no NO or NO₂ were detected. When the temperature was raised, no NO_x was detected until the temperature reached 50°C and no appreciable amount until 80°C. When the test was repeated at higher HNO₃ concentration (3 vs 1 M) and the purge gas was analyzed for CO, CO₂ and H₂ as well as NO_x, no reaction was seen until the temperature reached 80°C. These experiments demonstrated that no sugar-nitric acid reactions occurred below 50-80°C that produce gaseous products.

A second evaluation of the rate of sucrose nitration in SBW feed at ambient temperature was performed. In this test, WM-180 feed simulant from the Clemson-1 test was stored for roughly four months and spectrally analyzed to detect the presence of a C-O-N-O chemical bond using an FTIR instrument (entry A-23 in Table 2-1). Absence of such a bond would be evidence that nitration of the

^d Lotus Notes memorandum from N.R. Soelberg to K.J. Perry on 7/5/01 titled "Speciation of Iodine in offgas".

sugar (or any other organics present) had not occurred and this was, in fact, the case. This result provides assurance that a low temperature in the feed mixing tank during an abnormal shutdown would be sufficient to ensure that explosive products would not form by the action of HNO_3 on sugar in the solution.

DBE-139 Selection of Initial Baseline Process Configuration

A starting baseline process was selected by consensus in a group of technical personnel supporting the INEEL high level waste program (entry S-1 in Table 2-1). Additional possible process variations were later described and documented in Attachment 3.

DBE-144 Melter Feed Rate Basis

The melter feed rate will be determined by processing schedule requirements and melter sizing constraints. As part of an evaluation which determined the impacts of processing calcine in the SBW vitrification facility (Ref. 33), life-cycle costs were estimated for ten schedule/melter arrangements. The lowest cost arrangement of the ten that also completes SBW processing the soonest utilized a 4.5 m^2 melter.

Glass melt rates were determined for the RSM-1 and the Clemson-1 tests with values ranging from 7.1-11.1 lbm/hr-ft² for RSM-1 and an average rate of 3.0 lbm/hr-ft² for Clemson-1 (entries A-0 and G-1 in Table 2-1). These data are consistent with DWPF data that shows that a glass rate of about 6 lb/hr-ft² is as high as can be achieved over long operating periods.

Assuming the glass volume of the "Total SBW" mass balance (618 m^3 , Ref. 28), a glass rate of 6 lbm/hr-ft² (discounting the lower rate achieved in the EV-16 melter during the Clemson-1 test, and assuming the DWPF rate can be achieved with SBW), a 4.5 m^2 melter, and glass specific gravity of 2.6 the required operating time would be 12,200 hours, or 2.54 years at 200 operating days/year. To process the entire volume of SBW in 12,200 hours would require a feed rate of 79 gal/hr or 6.3 L/hr-ft². The range of raw SBW simulant feed rates achieved in the RSM-1 test was 8.6-13.5 L/hr-ft², utilizing the SBW-9 glass formulation at 30 wt% waste loading. Since the waste loading for SBW-22 is only two-thirds that of SBW-9 the equivalent range of processing rates using the SBW-22 formulation would be 5.7 to 9.0 L/hr-ft². Thus, the projected throughput requirement of 6.3 L/hr-ft² in the production facility is within the range that has been achieved.

DBE-149 NOxidizer Reduction Chamber Air Requirement

The air rate to the NOxidizer reduction chamber assumed in the current mass balance may be low by a factor of 7.2 (entry S-2 in Table 2-1). If this much additional air is required it will significantly increase the fuel requirement and hence the flow rate through the remaining sections of the NOxidizer and downstream equipment. Kinetic modeling suggests that the flow rate of NOxidizer effluent would increase by a factor 4.7 over the present mass balance. Such a large change suggests re-evaluating the selection of the NOxidizer relative to other NO_x abatement alternatives.

DBE-155 Slagging/Plugging Control at Melter Outlet

Pilot scale testing (Clemson-1) confirmed that solids accumulation in the melter offgas ducting and film cooler can be a significant problem (entry A-25 in Table 2-1). In this test solids obstructions in the offgas lines necessitated temporary shutdowns while to clear the offgas lines. Solids accumulation upstream of the film cooler and inside it were particularly problematic because the deposits were hard and difficult to remove, showing evidence of having been deposited in a molten state. As a result of

experience from this test it was recommended that (a) the melter exit duct should be located on the top of the melter, and maximally distant from glass and cold cap surfaces and from the feed stream to minimize the carryover of feed and solids, (b) a "bayonet" style melter should be inserted into the melter offgas exit with coolant injection immediately inside the film cooler inlet, (c) use of elbows in offgas ducts upstream of the quench/scrubber operations should be minimized and where needed, elbows should be "sweep" type (large radius of curvature), and (d) clean out ports in the off gas line upstream of the first quench would avoid the need to disassemble the piping during testing to remove deposits.

DBE-157 Control of Foaming in Melter

Foaming behavior was observed in a number of the tests conducted in FY 2001. In the Clemson-1 test only slight foaming was observed in the melt, and was apparently limited by use of a reducing agent (sugar at a concentration of 160 g/L) and by (unspecified) operating conditions (entry A-26 in Table 2-1). In melt rate tests at crucible scale foaming was so severe in batch type testing that a slurry-fed crucible test had to be developed whereby the feed was pumped into the crucible continuously (Ref. 7). In these incremental feeding tests foaming was observed at temperatures below 1000°C but not above (entry C-4 in Table 2-1). In research-scale testing of SBW-22 in RSM-2 excessive foaming was observed both in overly-reduced conditions (sugar concentration > 200 g/L) and overly-oxidized conditions (no sugar added—entry M-3 in Table 2-1). The first of these observations is consistent with glass formulation tests where it was noted that foaming intensity increased with sugar concentration during dryout of the simulated feed (entry E-6 in Table 2-1).

DBE-158 Speciation and Partitioning of Sulfate in Melter

An explanation of the observed behavior of sulfate in a melter feed was recently proposed^e. Alkali-nitrates, -phosphates, -sulfates, -chromates, -molybdates, -hydroxides, etc. form a mutually miscible molten salt as the first liquid phase to form during feed-to-glass reactions in the cold-cap. The salt is low viscosity and reacts readily with other waste and additive materials. Alkali-nitrates react releasing nitrogen-containing gases and intermediate phases or the first glass forming melt, while the alkali-sulfate, -phosphate, and -chromate remain in a molten salt phase after the -nitrate has all reacted. This remaining salt is known to segregate from the glass melt although the concentration of sulfur (and other salt components) in the melt is well below the solubility limit in the bulk glass at the melt temperature.

A possible sequence leading to segregation of the above-described salt layer is (a) an aqueous component segregates from the cold cap before the water evaporates, (b) the molten salt mixture forms and then drains away from other dried feed material as the water evaporates, (c) a molten sulfate salt phase stays behind after the nitrates decompose, (d) the sulfate salt layer remains segregated and establishes equilibrium with the final glass-forming melt, (e) the equilibrium is dynamic with sulfur migrating between the salt layer to and from local areas of the melt having different equilibrium sulfur concentrations.

In addition to being incorporated into the glass melt or segregating to a molten salt layer, sulfur in the feed may partition to the offgas by fuming of sulfuric acid or by reduction of SO_4^- to SO_2 . It may also precipitate from the glass melt as metal sulfides which eventually settle to the bottom of the melter. The fate of sulfur from the melter feed is of primary interest since both growth of a molten salt layer and precipitation of metal sulfides are unacceptable from an operations standpoint (see DBE-3 Detection of

^e This mechanism was discussed in a presentation by John Vienna of PNNL at the Tanks Focus Area coordination meeting held at the INEEL December 3-6, 2002.

Sulfate Salt Layer in the Melter). It has been found that the speciation and partitioning of sulfur during vitrification of SBW depends on a number of physical and chemical parameters. Among these are the sulfur content, reductant type and concentration, and acid and nitrate concentrations in the feed. In addition the feed rate, offgas flow rate (including sweep gas), and frit composition have been studied in various tests and found to influence the fate of feed sulfur to varying degrees.

Two glass formations were developed and tested during FY 2001 with the objective of preventing the formation and/or growth of a salt phase in the melter. These formulations were labeled SBW-9 and SBW-22. SBW-9 was selected from among several formulations on the basis of its having produced a salt-free glass (entry E-5 in Table 2-1). However, subsequent research-scale melter tests using this formulation were not salt-free (entries A-15, C-3, G-11 in Table 2-1). An additional concern with SBW-9 in these tests was an increase in the feed sulfur concentration over what had been assumed during initial tests of SBW-9 (see DBE-49 Waste Loading in Glass). The inability of the SBW-9 formulation to block salt formation led to testing and development of the SBW-22 formulation and adoption of a lower waste loading in the glass. Tests of SBW-22 have been completed in the SFMRF (at SRS) and the RSM (at PNNL). A partial test of SBW-22 was also performed in the EV-16 (Clemson). Preliminary results^f from these tests indicate that at 20% waste loading most of the sulfur in the feed was incorporated in the glass with no residual salt layer but with 5-10% of the feed sulfur still carried over to the scrub (entry M-16 of Table 2-1). The resulting sulfur content of the glass was approximately 1 wt% (as SO₃). The EV-16 melter was the largest scale tested to date but was abbreviated (due to the September terrorist attacks). Therefore, additional testing is needed at this scale (or larger) to fully demonstrate that the SBW-22 is truly "salt-free".

While the SBW-22 results demonstrate that partitioning of sulfur to a molten salt can be avoided by reducing the waste loading and possibly by tailoring of the frit, it has not been determined that this is the wisest course to pursue. For example, to reduce storage and disposal costs of the glass product it may be desirable to volatilize sulfur to SO₂ (and thus limit the *growth* of a salt layer) rather than attempt to dissolve all the feed sulfur in the glass and eliminate the salt layer entirely. Experimental studies conducted during FY 2001 have shed light on how different operating conditions and process inputs influence sulfur partitioning between the various phases discussed above. The information that was gleaned may be of use in directing further development of methods to control sulfate salt layer formation and/or growth in a melter without reducing waste loading.

In the initial CSM sulfur partitioning tests at PNNL with SBW-9 the only tests where *no* salt layer was observed were those using either 150% of the nominal sugar concentration or 240% of the nominal free acid (entry H-2 in Table 2-1). In these tests it was noted also that (a) the only test where 100% of the feed sulfur was dissolved in the glass was that where the reductant concentration was 0 g/L, (b) a 10-fold increase in the purge gas flow gave a slightly larger salt layer and reduced the %S retention in the glass from 73 to 67%, (c) with 67.5 g/L of sugar in the feed (50% of baseline) the %S retention increased from 73 to 96%, (d) doubling the S concentration in the feed resulted in a %S retention of 46% and a significant salt layer. Another significant result from this test was that nominal measured SO₂ concentrations were below 2 ppm in the offgas. When sufficient SO₂ was spiked into the sweep gas to yield a nominal 90 ppm in the offgas the measured SO₂ concentration of the offgas did not increase. This result, together with the fact that significant SO₂ concentrations in melter offgas were not observed in other tests (see entries G-11 and H-2 in Table 2-1), may indicate that volatilized sulfur from the melter (under the redox conditions tested) is present primarily in forms *other than* SO₂.

^f These results were reported by John Vienna at the Tanks Focus Area coordination at the INEEL on Dec 3-6, 2001. They had not been documented as of 2/5/02.

In the Clemson-1 test using the EV-16 melter the measured sulfur partitioning was approximately 57% to the glass, 12% to the scrub solution and solids, 28% to the gas phase, and 3% to the salt layer (entry A-15 in Table 2-1).

In summary, based on various FY 2001 testing, the following tentative conclusions may be drawn:

- (a) 0.8-1.0 wt% SO₃ appears to be the practical upper limit of sulfur concentration in glass from SBW;
- (b) Waste loading will have to be adjusted according to the concentration of sulfur in the feed if the objective is to confine *all* sulfur to the glass;
- (c) Even at 20% waste loading some sulfur will probably partition to the offgas and scrub liquor for WM-180 waste;
- (d) Addition of a reductant (sugar) to the feed reduces the %S retained by the glass and increases the %S which is volatilized;
- (e) From the data in hand we cannot yet define a set of conditions in a *large* melter which will *ensure* that a salt layer will either not form or (if formed) not grow beyond a prescribed extent (albeit the SBW-22 formulation in RSM-2 appears to have satisfied the former condition at a research-scale);
- (f) The trending data from testing to date could provide guidance in determining a set of conditions that will lead to i) absence of a salt layer and incorporation of most of the sulfur in the glass melt, ii) partitioning of most of the sulfur to the offgas (assuming either that scrub liquor is not recycled to the melter or that the majority of sulfur in the offgas is nonscrubbable and leaves the system in the gas phase), or iii) development of a steady state salt layer in the melter that can be tolerated. Whichever scenario is most acceptable from a risk and cost basis could then be pursued.

DBE-159 Feed Evaporation Requirement

In an engineering evaluation of feed evaporation, Wendt and Haefner (Ref. 22) determined that evaporating the liquid SBW to dryness would result in a savings in capital equipment of about \$1 million out of a total equipment cost of \$30 million, and would increase the glass rate by 50%. However, because of the uncertainties in how the evaporation would be done and whether the product would have acceptable properties, the study did not recommend changing the baseline flowsheet to include feed evaporation. Test data that addresses some of these uncertainties, together with several evaporation schemes, are discussed in the Attachment 3. The two drivers for reconsidering feed evaporation are cost and schedule. Attachment 3 indicates that evaporation, along with other pretreatment schemes, can offer significant savings in reducing the lifecycle cost and schedule for vitrifying SBW.

DBE-163 Fate of Chloride and Fluoride in the Melter

According to the mass balance reported for the Clemson-1 test, 98.5-100+% of Cl and F fed to the melter were found in the glass (entry A-10 in Table 2-1). However, the same test indicates 2.6 ppm Cl and 0.07 ppm F measured in the offgas (entry A-14 in Table 2-1) as well as in the sulfate salt taken from the melter (table 5.11 in Ref. 5). In light of significant over-recoveries of Cl and F in the glass, ranging from 131-459% and the fact that chloride concentrations in glass were not reported from other tests (except RSM-2 where some chloride measurements in the glass were obtained), the 98.5-100% figures must be viewed with some skepticism. Nonetheless, on the basis of the Clemson-1 data it is presumed that a high fraction of Cl and F would be retained in the glass and that the chloride in the melter offgas would be below the 75 ppmv MACT limit after suitable corrections for water vapor and oxygen (the

nominal correction to the 2.6 ppm total Cl in the offgas is roughly $1/(1-0.362) \times (11.5/7) = 2.6$ giving a value of 6.7 ppmv used for determining MACT compliance).

DBE-164 Should Sr-90 be Extracted from Scrubber Blowdown?

Removing ^{90}Sr from scrubber blowdown would permit disposal of the grouted blowdown waste as Hanford Category 1 (or NRC Class A) rather than Hanford Category 3 (or NRC Class C). For this reason Herbst, Kirkham and Losinski (entry P-1 in Table 2-1) recommend removing ^{90}Sr from the scrub. Whether ^{90}Sr should be removed depends largely on the conditions required for ^{137}Cs removal. Removal of strontium by ion exchange requires neutral or alkaline conditions. Test data is needed of the behavior of the scrub solution when it is neutralized and made alkaline. If heavy precipitation occurs, processing the solids may more than offset the savings from reduced disposal costs. If no precipitation occurs, the sorbent presently selected for cesium removal may also remove ^{90}Sr . Testing the cesium sorbent is needed to determine its performance at different levels of pH and sodium concentration in light of information from the cesium sorbent vendor which conflicts with historical test data from Hanford (Ref. 34) regarding the effect of sodium on the performance of the cesium sorbent.

DBE-167 Soluble Species in Off-gas

Information regarding soluble species in the offgas can be gleaned from analysis of scrub solids and solutions from Clemson-1 (entry A-18 of Table 2-1) and RSM-2 (entries M-10 through M-14 and M-17 of Table 2-1). Based on analysis of scrub solid and liquid samples from Clemson-2 tests, dissolved species include B, Cs, Li, K, SO_4 and Na, while species found in the solid phase include Al, Ca, Ce, Cu, Fe, P, Si, and Zn. Ca, Fe and Zn partition to solids and liquid in approximately a 3:1 ratio, Cr in 1:2 ratio, Cu and Ni in a 3:2 ratio, and Ru in a 1:1 ratio. XRD analyses of solids filtered from the scrub solution in the Clemson-1 tests indicate that major compounds include SiO_2 , K_2NaAlF_6 , and CaF_2 . The data thus indicate that K, Na, Ca, Fe, Zn, Cr, Cu, Ni, and Ru form both soluble and insoluble species in the offgas.

Samples of scrub liquid and undissolved solids filtered from the scrub were analyzed in RSM-2 tests. Soluble species included B, Ca, Cl, K, Na and SO_4 , and insoluble species included Fe, Si, Ti, and Zr.

DBE-174 NOxidizer Reduction Chamber Chemistry Using Propane

NOxidizer tests at MSE (entry J-1 of Table 2-1) demonstrated 94-96% NO_x destruction efficiency from a simulated offgas containing up to 40,000 ppm NO_x and a NOxidizer fueled by propane. A matrix of thirteen tests was designed to determine the effects of three parameters – T_{quench} , $(T_{\text{reoxidation}} - T_{\text{quench}})$, and $[\text{O}_2]_{\text{exit}}$ (T = temperature, $[\]$ = concentration) – on NO_x destruction efficiency. The highest NO_x destruction efficiencies were seen with (a) low quench temperature (1600°F), low temperature difference (50°F) and high oxygen (>4%), (b) low quench temperature (1600°F), high temperature difference (200°F) and low oxygen (1.5%), and high quench temperature (1800°F), low temperature difference (50°F) and low oxygen (1.5%).

Kinetic modeling (see entry Q-1 of Table 2-1 and Ref. 21) provides more detail of the reduction chamber reactions, although this modeling was based on natural gas rather than propane because of the availability of kinetic mechanisms for natural gas. Based on 4.96% NO_x in the NOxidizer feed, the model shows 99.94% NO_2 destruction in the reduction stage, 85% NO destruction in the reduction stage, 14% destruction of NO in the quench and reoxidation stages, and a slight gain of NO_2 in the quench and reoxidation stages such that the overall destruction NO_2 efficiency is 99.95%.

4. STATUS OF SBW VITRIFICATION PROCESS

During FY 2001, a baseline flowsheet for vitrification of SBW was defined, and a database (the TBDB) was developed and populated that contains a mass balance for the flowsheet, the basis for the mass balance, and plans for obtaining data needed to validate the flowsheet basis. The flowsheet consists of three primary systems – the feed system, the melter and the offgas treatment system. This section examines the FY 2001 work from a process viewpoint in contrast to the low level discussion of the individual DBEs in Section 3. In this higher level context, the following subsections summarize the FY 2001 work in light of (a) progress toward obtaining the data needed for each of the three primary systems listed above, (b) insights gained that suggest additional development, not envisioned in the present database, is needed, and (c) major data needs that have not been addressed.

4.1 Feed System

The major equipment in the feed system includes (a) an SBW storage tank and pumps, (b) a sugar solution storage tank and pumps, (c) glass frit storage and transfer equipment, (d) melter feed mix tanks and associated pumps and (e) melter feed tanks and associated pumps. The function of the feed system is to deliver a homogeneous feed of waste, glass forming components and reductant to the melter at the rate required to meet the treatment schedule.

4.1.1 Feed Rheology, Feed Reactions and Frit Versus Glass Forming Components

Initial feed rheology studies found no gelling or caking to occur during feed mixing, and RSM-1 and other melter tests experienced no feed line plugging during the tests. No settling or segregation of feed solids was noted in any test. Thus, within the parameters of the tests, i.e., WM-180 simulant and glass forming components equivalent to a waste loading of about 35%, transfer of a homogeneous feed to the melter has been verified. However, tests of the RSM mixing and feed system are needed at the new baseline waste loading of 20%. Use of alkaline glass forming components such as CaO and Li₂O resulted in neutralization of the SBW, precipitation of solid and gelatinous species, and the release of heat during the RSM-2 test. In this test, which used a 20% waste loading and the SBW-22 formulation with higher alkaline fraction, additional acid was needed in the feed in order to prevent gelling.

The rheology studies found the melter feed to be very abrasive and chemically aggressive; these results will dictate material design requirements for the transfer, mix and storage equipment. An evaluation of several forms of silica resulted in the recommendation to use –400 mesh silica.

No evaluation of frit versus glass forming chemicals has yet been performed. This evaluation is needed to further define the number and size of storage vessels for these materials, to better define feed rheology, to better determine temperature and chemical control requirements of feed and mix tanks, and to confirm glass melt rate, glass properties and other characteristics of the glass melt and melter operation.

It was shown (Ref. 18) that no reaction between sugar and nitric acid in the waste occurs below 50°C. Confirmation tests of this limiting temperature are needed to establish for the full scale process the temperature and cooling requirements of the melter feed mix tanks and melter feed tanks.

4.1.2 Frit Formulation and Feed Mixing

Frit formulation relates primarily to glass quality and will be discussed in Section 4.2.1; however, the frit formulation(s) sets requirements for the feed system. Progress has been made determining glass property/composition relationships through CVS work. These relationships should be used to determine

whether a single or multiple frit formulations would be required to process waste from the different INTEC Tank Farm tanks. The CVS models should also be used to evaluate potential benefits of feed mixing (i.e., pre-blending the liquids from all SBW tanks in the tank farm) relative to glass formulation; i.e. reduced glass volume, a single frit formulation, or increased margin in the envelope of feed composition.

4.1.3 Reductant Type and Form

Sugar was shown to be the only reductant, of four tested, which met requirements for not affecting glass product properties (entry C-5 in Table 2-1), being able to reduce iron in the melter without final inclusion in the glass, and not adversely affecting the feed transport properties. In contrast to the baseline assumption of 67 wt % sugar solution added to SBW, all melt tests to date have used granular sugar as the reductant. As discussed more fully in the Attachment 3, use of granular sugar has advantages of reducing the amount of water in the feed and avoiding the need to heat the sugar syrup to keep it pumpable, and is consistent with the test data. Thus granular sugar is recommended.

4.1.4 Feed Pretreatment

The baseline flowsheet assumes no feed pretreatment. However, one of the major results of this past year's development testing is the finding that the maximum glass waste loading is only about 20%, limited by the amount of sulfate in the waste feed. Tests this year have also demonstrated that >80% of the sulfate could be precipitated from an SBW simulant by the addition of barium nitrate, and that the resulting precipitate could be separated from the liquid waste (Ref. 13). Peeler and Vienna (Ref. 9) estimate that if the waste sulfate concentration did not limit waste loading, the glass formulation would be constrained by the sodium content and the waste loading would increase to 38%. Thus sulfate precipitation offers a means of greatly reducing the volume of glass by nearly a factor of 2 with associated savings in equipment costs, glass storage facility costs, and glass disposal costs.

Other feed pretreatment steps offer additional process and cost benefits. The total melter feed is approximately 50% water, and the melter offgas is more than 70% H₂O. Evaporating the feed to dryness would reduce the melter offgas by a factor of more than 3, resulting in significant savings by downsizing the needed offgas equipment. In addition the melt rate would double (Ref. 22). Though no evaporation tests of SBW were performed in the past year results from earlier tests provide a preliminary basis for evaluation of evaporation schemes. These studies are cited and evaporation schemes discussed in Attachment 3.

A third pretreatment scheme which offers potential cost savings is to remove nitrates by crystallization. Combined with sulfate precipitation, this scheme has the potential of reducing the feed rate by 91% and the product glass volume by 70% (see Attachment 3).

The potential savings offered by pretreatment schemes provide incentive for further evaluation and testing of these schemes.

4.2 Melter and Canister Filling System

The function of the melter and canister filling system is to produce from the feed waste a vitrified product that meets all disposal requirements. Progress was made in the past year determining the limiting waste loading in the glass, developing glass formulations that result in glass meeting both operating and disposal requirements, and characterizing the melter offgas. However, all data obtained to date is strictly applicable only to a WM-180 simulant waste feed.

4.2.1 Glass Waste Loading, Glass Formulation and Control of Salt Layer

Glass formulation studies in the first half of the year (Ref. 9) resulted in the recommendation of a formulation called "SBW-9" and a waste loading of 30 wt%. Using a formulation very similar to SBW-9 and varying waste loadings between 30% and 35%, RSM-1 tests produced glasses that met all processing and disposal requirements. However, in tests performed in the EV-16 melter at the Clemson Environmental Laboratory using the SBW-9 formulation and a waste loading of 30%, a salt layer was observed on the melt during the demonstration. This result, along with an updated WM-180 surrogate formulation (Ref. 25) that showed a 40% higher sulfate concentration in the waste, led to development of additional glass formulations. The objective of these later formulations was to incorporate all feed sulfate in the glass. To achieve this objective a new glass formulation ("SBW-22") and a waste loading of 20% were recommended. The RSM-2 tests, which used the higher-sulfate waste formulation, the SBW-22 formulation, and the recommended 20% waste loading resulted in no salt layer accumulation and yielded glass with acceptable properties.

The tests performed this past year thus showed that for WM-180 waste, with a sulfate concentration of 0.07 M, limiting the waste loading to 20% (or less) is one method to control salt layer accumulation. A glass formulation was developed that captured >90% of the feed sulfate in the glass. However, several issues regarding waste loading remain. Glass formulations for wastes other than WM-180 have not been developed. Schemes have been proposed to reduce the sulfate concentration in the feed and hence permit higher waste loadings, i.e., feed blending and feed pretreatment, but surrogates for the waste from these schemes have not been tested. Feed blending would reduce the sulfate concentration in the waste to the level of the surrogate in tests early in the year, for which a waste loading of 30% was recommended. Feed pretreatment has the potential of increasing the waste loading to the 35-40% range. Another alternative that has been discussed to increase waste loading is to allow a limited sulfate salt layer to develop and restrict its growth by volatilizing sulfur. Some of the test data suggests that such an approach may be feasible. Based on the sizable economic benefits from increasing the waste loading, further consideration of these alternative process schemes is recommended.

4.2.2 Melter Scale-Up

Melter tests were performed in the past year in the EV-16 melter, the Research-Scale Melter (RSM), the slurry fed melt rate furnace (SFMRF) and the centimeter-scale melter (CSM). The EV-16 melter has a chamber 18-in by 18-in, with a design depth of 16-in. The nominal glass volume in the EV-16 melter is 3 ft³. The RSM chamber has a diameter of 6-in and a nominal glass depth of 3-in, equating to a glass volume of about 0.05 ft³. The SFMRF and CSM both utilize crucibles. The SFMRF crucible typically contains 5 kg or 0.07 ft³ of glass. The CSM tests with 1-1.5 inch diameter melt surfaces are typically used to obtain between 10 and 50 gm of glass.

Past experience at other DOE vitrification facilities as well as budget limitations has dictated the use of different subscale melters to obtain data for the design of vitrification facilities for INEEL waste. While the data obtained has been useful in preparing mass balances, feasibility designs and initial cost estimates, future tests will be needed to determine the effects of scale on melter performance parameters. Knowledge of such will be necessary before a full size design can be formulated with confidence.

4.2.3 Melter Operating Parameters

Glass melt rates determined from the RSM-1 and the EV-16 tests were generally consistent with data from existing, full-scale melters at DOE sites, and provide a basis for determining the required melter size for a given treatment schedule or processing schedule for a given melter size. The melt rate for a dried feed should be determined as part of an evaluation of pretreatment schemes.

The melter plenum temperature has a significant impact on offgas equipment sizing requirements, since the melter offgas must be cooled by adding air and steam in the film cooler. The RSM-1 tests were run with plenum temperatures significantly lower than presently assumed in the baseline mass balance (536°C compared to 650°C), and the Clemson-1 tests included periods targeting a plenum outlet temperature of 450°C with no apparent deleterious effects. Additional offgas, melt rate, and processability data are needed to establish the minimum plenum temperature required to allay flammability, melt viscosity, and throughput concerns. However, based on the advantages of operating at lower temperature and the offgas data from the Clemson-1 test, it is recommended that the baseline plenum temperature be changed to 500°C.

4.2.4 Melter Partition Factors and Reactions

Data was obtained from RSM-1, RSM-2 and Clemson-1 tests relative to partitioning of semivolatile and nonvolatile species between glass and offgas. Off-gas data from these tests also provides a basis for mass balance assumptions regarding reactions that occur in the melter, producing CO, CO₂, NO, NO₂, SO₂, HCl, HF, HI and H₂. Partitioning data is needed to determine the requirements for offgas treatment, to confirm that the baseline process flow scheme will produce secondary wastes that can be disposed and offgas that meets emission standards, and to evaluate options for processing scrub wastes.

The partition factor data have large uncertainties. Part of the uncertainty is due to incomplete mass balance closure. For example, based on the steady state portion of the Clemson-1 test, 98.5% of the nominal Cl in the feed was found in the glass, and 32% was found in the scrub solution. The partition factor (fraction of input mass leaving the offgas) for Cl based on the scrub would thus be 32%, but based on the glass it would be only 1.5%. For some species, the concentration found in the offgas is near or below the detection limit, and use of a detection limit to calculate a species partition factor can introduce large uncertainty. Finally, the effect of scale or melter configuration on the partition factors is unknown and result in misleading results. For example, the DF (inverse of the partition factor) for cesium was found to be 1.5-4.9 in the Clemson-1 tests, 3-5 in the RSM-1 test, and 9-49 in the RSM-2 test. However, data for large-scale melters typically shows Cs DFs of 80-100 (for example, see Ref. 32).

In spite of the large uncertainties in the data, the database should be updated to reflect findings from the RSM and Clemson tests using the most conservative values relative to offgas system design. The numeric changes to specific factors may depend on the direction of future mass balances, but the following should be considered:

- SO₃ in the offgas would be absorbed by quench and scrub solutions producing H₂SO₄, affecting the acidity of the solutions which in turn will determine their effectiveness in removing other species from the offgas. Measurable SO₃ was seen in the RSM-2 data.
- Considerably less SO₂ was seen in the RSM-2 offgas than the other melter tests, and may be because of the use of the SBW-22 formulation of GFCs, although extremely low levels of SO₂ were also noted in CSM testing with other formulations. Based on RSM-2 results indicating 5-10% of sulfur from the feed was recovered in the scrub, the extent of reaction of sulfate to SO₂ in the mass balance model should be reduced from 14% to around 6%.
- Concentrations of HI and I₂ were measured in RSM-2 melter offgas. The quality of this data should be assessed. Depending on the results of this assessment it may be appropriate to use the RSM-2 HI and I₂ concentrations as a basis for specifying reactions producing these species in the mass balance model. The model presently assumes all I in the offgas is present as NaI.

- Appreciable concentrations of N_2O were seen in some tests. Appropriate reactions producing N_2O should be added to the mass balance model.
- Partition factors for semi-volatile and nonvolatile species in the model should be updated based on the recent data. However, for species with poor mass balance closure or poor agreement with DWPF and other large-scale data, partition factors based on large-scale data may be preferable.
- Particle-size distribution data from Clemson-1 offgas solids can be used to refine calculations of solids scrubbing efficiencies used in the mass balance until better PSD data is obtained.
- Concentrations of HNO_3 in the melter offgas were measured in the Clemson-1 tests, and should be used as a basis for volatilization of HNO_3 in the melter in future mass balances.
- Appreciable concentrations of Cl_2 were seen in some tests. Appropriate reactions producing Cl_2 should be added to the mass balance model.

4.3 Off-Gas Treatment System

4.3.1 Film Cooler

The function of the film cooler is to prevent deposition of solids in the offgas exiting the melter plenum in offgas piping. Air and/or steam is added in the film cooler to cool the offgas and entrained solids to a temperature below that which solids would stick to walls of piping and equipment. Appreciable deposition of solids in offgas piping was seen in the Clemson-1 tests with blockage in a duct connecting the melter plenum to the film cooler. To avoid this the film cooler needs to be close-coupled with (i.e., inserted into) the melter offgas exit. Deposits downstream of the film cooler were generally of a friable, powder-like consistency, and adhered very loosely to the pipe walls. Additional tests of alternative configurations (e.g., smaller piping to increase gas velocity) are needed to demonstrate a design that adequately prevents solids deposition in offgas piping.

4.3.2 Off-gas Quench and Scrub System

The primary functions of the offgas quench and scrub system are to cool the offgas to saturation temperature, to remove radionuclides to the extent that further processing of the offgas can be done out of cell, and to process or treat the resulting scrub solution and solids.

4.3.2.1 Off-gas Quench and Scrub

The offgas treatment system of RSM and Clemson downstream of the film cooler differs from the baseline flowsheet. The RSM system includes an ejector venturi scrubber (EVS) and high efficiency mist eliminator (HEME), the Clemson system includes a quench chamber, air atomized scrubber, cyclone separator, packed bed and mist eliminator. The baseline flowsheet has a quench column, venturi scrubber and HEME. Considerable review and evaluation of offgas treatment systems has been performed in the past year (Refs. 19, 23). Final selection of unit operations, particularly for offgas scrubbing, is dependent on better definition of the contaminants in the melter offgas. Data gained in the past year contributes to filling this data gap.

4.3.2.2 Soluble and Insoluble Species

The baseline flowsheet includes separation of solids captured in the quench and scrub liquids, with recycle of these solids to the melter. The mass balance assumes chemical species are either totally soluble

or insoluble. Information regarding soluble and insoluble species in the scrub was obtained from analysis of scrub solids and solutions from Clemson-1 (Ref. 5) and RSM-2 (Ref. 17) tests. These analyses should be used to update the baseline lists of soluble and insoluble species (see discussion for DBE 167).

Besides shifting some species from one group, soluble or insoluble, to the other, the scrub solids and liquid analyses along with XRD analyses indicate that some elemental species partition between both soluble and insoluble solids. Sensitivity studies are needed to determine the impacts of species partitioning between soluble and insoluble solids. These studies will determine whether additional test data should be obtained to better determine partition factors, the range of these factors, and how different process parameters such as melter plenum temperature or scrub pH affect the partition factors.

4.3.2.3 Scrub Liquids and Solids Treatment

The baseline system of scrub treatment includes several untested systems, including solids separation, pH adjustment of scrub purge, removal of cesium by ion exchange, and grouting of the cesium-free scrub purge. In addition, recycle scrub has not been included in the feed simulant used in any melter test to date. Insights gained from experiments this past year include the following:

- Data from Clemson-1 showed that a large fraction (>50%) of the fraction of solids collected in the scrub are less than 1 micron in diameter. This finding would preclude separation of solids by settling.
- Analysis of RSM-2 samples, the only melter test which included mercury in the feed, showed no mercury in the HEME effluent gas. Experiments that determined mercury VLE data at scrub conditions confirmed that a very high fraction of mercury in the melter offgas would be captured by the scrub. While questions remain regarding the speciation of mercury in the offgas and scrub, the data suggests that higher-than-expected mercury scrubbing efficiency may necessitate removal of mercury from the scrub.
- Undocumented tests of neutralization of simulated scrub solution showed significant precipitation. Assuming this was not an artifact of the test procedure it suggests that filtration will be required upstream of the cesium ion exchange column. Further testing is needed to establish that the observed precipitation is representative of the actual scrub liquid. Analysis of the composition and quantity of solids should be performed to determine if they could be recycled to the melter or require a different disposal path.
- Although no tests were performed this past year of the proposed cesium sorbent, some data was recently received from earlier tests performed at Hanford on a similar waste stream. The data shows that increasing sodium concentration in the waste decreases cesium removal efficiency. The sodium concentration expected in the SBW vitrification plant scrub solution is much higher than the waste tested at Hanford.
- Significant levels of organics (~1 g/l) were seen in the scrub solution of RSM-2 tests. The effect of these organics on grouting the scrub purge will need to be determined.

Partly because of these test results and partly because of identified uncertainties in the baseline scrub treatment system, modifications and alternative processes have been proposed for scrub treatment. These include:

- Adding steps to remove mercury by electrochemical reduction and amalgamate the resulting elemental mercury to the baseline scrub treatment system.

- Adding a step to precipitate mercury from the scrub purge.
- Removing strontium in addition to cesium from the scrub purge.
- Using a moving bed with continual replacement of sorbent to remove cesium and continual bleed of the loaded sorbent to the melter.
- Treating the scrub purge by evaporation to produce a dry solid waste. Purge from the downstream caustic scrubber would also be treated by evaporation. This alternative would eliminate recycle of scrub to the melter, ion exchange to remove cesium (and strontium), and the grouting step.

An evaluation of scrub treatment alternatives is needed, followed by testing to confirm the performance of those that offer the greatest benefits.

4.3.3 Off-gas Solids Filtration

The evaluation of secondary wastes (Ref. 20) recommended replacing the in-cell HEPA filters with washable HEPA filters. Some development testing of washable filters has been performed at SRS. Review of the data from this program is recommended. It may be desirable to commission additional parametric testing to determine the feasibility of washable filters in the SBW vitrification process. The alternative to washable filters is the conventional fabric HEPA filters that must be treated for disposal in the NWCF filter leach facility.

4.3.4 NO_x Abatement

Requirements for NO_x abatement were identified (Ref. 23) and calculations performed to estimate the maximum allowable NO_x concentrations at the stack. Tests have confirmed that the NO_x oxidizer will reduce effluent NO_x concentrations well below the allowable limit (Ref. 14).

Information from the NO_x oxidizer vendor, however, shows that a much higher rate of air to the reduction chamber will be required than is presently assumed. Based on the higher air rate, the NO_x oxidizer effluent rate will be 6-8 times the inlet rate, compared to the baseline mass balance of 1.7 times. The increased offgas volume will have significant implications for the size of downstream equipment and also the volume of spent carbon waste. Given that alternative NO_x abatement technologies were ranked almost equal to the NO_x oxidizer in past NO_x abatement technology evaluations (Ref. 23), testing of SCR technology is recommended.

4.3.5 Acid Gas Removal

The caustic quench and scrub step cools the NO_x oxidizer effluent and removes acid gases, including HCl, HF, HI, and SO₂. No data was obtained in melter tests this past year regarding offgas compositions downstream of scrubbing units. As discussed in Attachment 3, the need for acid gas removal has not been established. Thus it is recommended that in future melter tests, samples of gas downstream of HEPA filters be taken and analyzed in order to determine whether additional acid gas removal would be required.

4.3.6 Mercury Removal

Tests were performed using small-scale GAC beds (0.4-2 cm³) to determine the effects of feed gas composition, form of mercury and residence time on mercury removal efficiency (Ref. 6). While the

effluent concentrations from the tests were within MACT emissions limits, the inlet concentrations were approximately 100 times less than the current mass balance concentration. Additional melter tests that include mercury in the feed are needed to better establish the mercury concentration downstream of scrubbing equipment. Then, additional testing at appropriate feed concentrations and a larger range of residence times is needed to clearly determine the effects of inlet mercury speciation and residence time on mercury removal efficiency.

5. REFERENCES

1. Consent Order and Settlement Agreement between DOE and the State of Idaho Regarding Spent Nuclear Fuel and Nuclear Waste Issues, October 17, 1995.
2. DOE, Idaho High-Level Waste & Facilities Disposition Draft Environmental Impact Statement, DOE/EIS-0287D, December 1999.
3. Tank Focus Area, *Assessment of Selected Technologies for the Treatment of Idaho Tank Waste and Calcine*, PNNL-13268, August 2000.
4. A. Olson, J. Murphy, K. Perry, *Pre-Decisional Sodium Bearing Waste Technology Development Roadmap*, INEEL/EXT-2000-01299, September 2000.
5. K. J. Perry, R. R. Kimmitt, N. R. Soelberg, R. D. Tillotson, A. N. Olson, *Test Results from SBW-FY-01-PS-01 Vittrification Demonstration of Sodium Bearing Waste Simulant Using WM-180 Surrogate*, INEEL/EXT-01-01073, August 2001.
6. J. A. Del Debbio, T.T. Watson, R. J. Kirkham, *Removal of Mercury from SBW Vittrification Off-Gas by Activated Carbon*, INEEL/EXT-01-01227, September, 2001.
7. J. A. McCray, D. L. Griffith, R. R. Kimmitt, D. D. Siemer, *Status of Melt Rate Testing and Reductant Selection for SBW Vittrification*, INEEL/EXT-01-01223, September, 2001.
8. L. G. Olson, *Radioactive Crucible Scale Glass Melts Using INTEC Tank WM-180 Sodium-Bearing Waste*, INEEL/EXT-01-01020, September, 2001.
9. D. K. Peeler, *Glass Formulation Development for INEEL Sodium-Bearing Waste (FY2001 WM-189)*, WSRC-TR-2001-00295, September 21, 2001.
10. D. W. Marshall, *FY-2001 Accomplishments in Off-Gas Treatment Technology Development*, INEEL/EXT-01-01302, September, 2001.
11. R. W. Goles, J. M. Perez, B. D. MacIsaac, D. D. Siemer, J. A. McCray, *Test Summary Report INEEL Sodium-Bearing Waste Vittrification Demonstration RSM-01-I*, PNNL-13522, May, 2001.
12. J. G. Darab, D. D. Graham, B. D. MacIsaac, R. L. Russell, H. D. Smith, J. D. Vienna, D. K. Peeler, *Sulfur Partitioning During Vittrification of INEEL Sodium Bearing Waste: Status Report*, PNNL-13588, July, 2001.
13. C. M. Nitzel, *Separation of Barium Sulfate Precipitate from Liquid Sodium-Bearing Waste Simulant*, August, 2001.
14. MSE Technology Applications, Inc., *John Zink NOxidizer Propane-Fired Performance Assessment and Mercury Emission Speciation*, PTP-81, February, 2001.
15. J. A. Nenni, *ETS Process Parameter and Outlet Stream Predictions for WM-185 Feed*, INEEL Interoffice Memorandum, January 9, 2002.
16. R. J. Kirkham, *Mercury Scrubbing Vapor Liquid Equilibrium* (unpublished report)

17. R. W. Goles, J. A. Del Debbio, R. J. Kirkham, B. D. MacIsaac, J. A. McCray, D. D. Siemer, N. R. Soelberg, *Test Summary Report INEEL Sodium-Bearing Waste Vitrification Demonstration RSM-01-2*, PNNL-13869, May, 2002.
18. R. Russell, H. Smith, *Summary Report for Rheology Testing of INEEL Sodium Bearing Waste (SBW) Simulant and Melter Feed*, March, 2001.
19. D. R. Marshall et al, *Evaluation of Scrubber Operations for Treatment of Idaho Waste Vitrification Facility Off-gas*, INEEL/EXT-01-00952, September, 2001.
20. A. K. Herbst, R. J. Kirkham, S. J. Losinski, *Secondary Waste Considerations for Vitrification of Sodium Bearing Waste at the Idaho Nuclear Technology and Engineering Center FY-2001 Status Report*, INEEL/EXT-01-01172, September, 2001.
21. R. A. Wood, *Modeling NO_x Destruction Options for INEEL Sodium Bearing Waste Vitrification*, INEEL/EXT-01-01220, September, 2001.
22. D. S. Wendt, D. R. Haefner, *Wet vs. Dry SBW Vitrification Feed Evaluation*, EDF-2860, October 11, 2001.
23. R. A. Wood, D. Tyson, B. Bonnema, C. Olsen, A. P. Pinto, D. Wendt, S. Reese, B. Raivo, *Feasibility Study for the Idaho Waste Vitrification Facilities Off-gas Treatment for Sodium-Bearing Waste*, INEEL/EXT-01-00995.
24. C. M. Barnes, *Feed Composition for the Sodium-Bearing Waste Treatment Process*, INEEL/EXT-2000-01378, Revision 1, July, 2001.
25. J. D. Christian, *Composition and Simulation of Tank WM-180 Sodium-Bearing Waste at the Idaho Nuclear Technology and Engineering Center*, INEEL/EXT-2001-00600, May, 2001.
26. J. D. Christian, *"Worst Case" Simulant for INTEC Sodium-Bearing Waste Vitrification Tests*, INEEL/EXT-01-01219.
27. N. R. Soelberg, et al, *Sodium Bearing Waste Vitrification Test Results*, paper presented at IT3'02 Conference, May 13-17, 2002, New Orleans, LA.
28. Dean Taylor, Charles Barnes, Lance Lauerhass, *INEEL SBW Vitrification Process*, INEEL/EXT-01-01139, September 2001.
29. J. D. Christian, *Technical Baseline Database for SBW Vitrification: Fluoride Concentration Requiring Use of Additives for Corrosion Control*, INEEL Interoffice Memorandum to D. D. Taylor, September 24, 2001, JDC-04-01.
30. R. R. Hammer, *A Determination of the Stability Constants of a Number of Metal Fluoride Complexes and Their Rates of Formation*, ENICO-1004, August, 1979.
31. J. A. Murphy, *Determination of the Zirconium Fluoride Stability Constants by Direct Measurement of Equilibrium Hydrofluoric Acid Using the Amperometric Response of Titanium and Hafnium Electrodes*, WINCO-1098, May 1992.
32. N. E. Bibler, T. L. Fellingner, S. L. Marra, R. J. O'Driscoll, J. W. Ray, *Tc-99 and Cs-137 Volatility from the DWPF Production Melter during Vitrification of the First Macrobatches at the Savannah River Site*,

33. S. O. Bates, B. D. Raivo, J. J. Quigley, S. M. Berry, W. H. Landman, S. L. Palmer, T. M. Hipp, *Feasibility Study for Vitrification of Calcine in the Idaho Waste Vitrification Facility*, INEEL/EXT-01-00978, September, 2001.
34. L. A. Bray, K. J. Carson, R. J. Elovick, D. E. Eakin, *HWVP Transuranic Process Waste Treatment by Ion Exchange*, HWVP-90-1.2.2.04.04A, February, 1990.
35. C. M. Barnes, *Feed Composition for the Sodium-Bearing Waste Treatment Process*, INEEL/EXT-2000-01378, Revision 0, October 2000.

ATTACHMENT 1

Table A-1: Complete list of Design Basis Elements (DBEs) in technical baseline for SBW vitrification.

DBE ID	DBE Name
2	Fate of sulfur in NOx abatement
3	Detection of sulfate salt layer in the melter
15	Disposition/handling of noble metals in melter
16	Submicron particle size distribution out film cooler
17	Nitrate destruction chemistry in melter
18	Melter chemistry
19	Target concentration of total dissolved solids (TDS) in melter feed
20	Reductant selection and concentration in melter feed
21	No secondary liquid waste streams
22	Pumping behavior of melter feed slurry
23	Flammability limits on H ₂ , CO, and unburned HCs in offgas
24	Partition factors for feed species in melter
25	Iodine-129 concentration in SBW feed
26	Volatilities of radionuclides (excluding Cs, Tc-99, and I-129) in melter
27	Partitioning of acid gases and Hg during quench/scrubbing operations
28	Dioxins/Furans in offgas system
29	Carryover of melter feed to offgas
30	Representativeness of cold simulants in melter tests
31	Composition envelope for acceptable glass feed
32	What tank farm solids will be coprocessed with SBW?
34	Frit vs GFC in melter feed
35	SBW feed blending
36	Evaporation limit for SBW
37	Total volume and composite SBW feed composition
38	Noble metals concentration in SBW feed
39	Corrosion of melter components
40	Canister fill height measurement
41	Canister closure method
43	Composition and concentration of solids in SBW feed
44	Required control of feed mix tank temperature
45	Feed preparation requirements (excl. evaporation)
46	Steam/air ratio in film cooler
47	Melter offgas temperature
48	Glass frit composition
49	Waste loading in glass
50	Melter air rate
51	Processing pressure in melter
52	Pressure drops through offgas system components
53	NOxidizer reduction chamber chemistry using kerosene
54	NOxidizer oxidation chamber chemistry
55	NOxidizer reduction chamber reaction and quench temperatures
56	NOxidizer refractory changeout frequency
57	Disposal path for spent NOxidizer refractory
58	Composition of spent NOxidizer refractory
59	Speciation of mercury in NOxidizer effluent

DBE ID	DBE Name
60	pH control in scrubber solutions
62	Off-gas system corrosion control
63	Off-gas flow control system
64	Air addition to NOxidizer oxidation chamber
65	Mercury retention and speciation in the melter
66	Mercury speciation downstream of melter
67	Peak acceptable mercury loading in GAC
68	Residence time requirement for Hg extraction in GAC
69	Maintenance and reliability issues for GAC columns
70	Mercury removal efficiency of GAC beds
71	Maximum allowable NOx concentration in GAC bed influent
72	Process requirements for sulfide precipitation of Hg
73	Electrolytic reduction of oxidized mercury
74	Solids scrubbing efficiency in acid quench
75	Solids scrubbing efficiency in acid venturi scrubber
76	Solids scrubbing efficiency in caustic quench
77	Species removal efficiencies in knockout drum
78	Species removal efficiencies in HEME
79	Species removal efficiencies in upstream HEPA filter bank
80	Species removal efficiencies in downstream HEPA filter bank
81	Effects of 'other' species on Hg polishing effectiveness of GAC
82	Removal efficiencies for dioxins/furans in GAC column
83	Maximum acceptable undissolved solids content of scrubbing solutions
84	Maximum acceptable dissolved solids content of scrubbing solutions
85	Maximum allowable Cl, F content of scrubbing solutions
86	Fluoride concentration requiring use of additives for corrosion control
87	Required corrosion control additives
88	Particulate removal efficiency in WESP
89	Hg removal efficiency in WESP
90	NOx removal efficiency in WESP
91	Liquid/gas ratio in scrubbers and quench operations
92	Catalytic oxidizer operating parameters
93	Scrubbing temperatures
94	Solids removal efficiency in demister
95	Loading of Cs in ion exchange resin
96	Choice of target disposal site for grouted waste
97	Waste loading of scrubber blowdown in grout
98	Grout formulation for scrubber blowdown
99	Thermochemical heat release during grouting of scrubber blowdown
100	Maximum allowable solids loading in HEPA filters
101	Settling rate of solids in scrubber blowdown liquid
102	Will facility comply with MACT
103	Glass canister size
104	Waste acceptance criteria for mixed waste disposal at WIPP
105	Waste acceptance criteria for waste disposal at Yucca Mountain
106	Waste acceptance criteria for mixed waste disposal at NTS

DBE ID	DBE Name
107	Waste acceptance criteria for mixed waste disposal at Envirocare
108	Waste acceptance criteria for mixed waste disposal at Hanford
109	Glass durability requirement
110	Glass homogeneity requirement
111	Glass cooling requirement
112	Melter feed homogenization requirement
113	Target disposal sites for secondary wastes
114	Melter selection
115	Choice of melter pilot test facility
117	Target disposal site for glass
118	Disposal requirements for mercury-containing secondary wastes
120	NOx abatement requirement in process offgas
122	Radioactive process streams to be recycled to melter
123	Should Cs-137 be extracted from scrubber blowdown
124	Glass formulation for scrub purge ion exchange sorbents
125	Disposal path for spent HEPA filters
126	Disposal path for GAC
127	Selection and performance of Cs-137 ion exchange sorbent
128	Composition of ion exchange sorbent
129	Preheating requirements for HEPA filters
130	GAC TCLP performance
131	H2 generation mechanisms in grout
132	Gas generation in Cs sorbent
133	Speciation of iodine in melter offgas
134	Destruction/removal efficiency (DRE) for ammonia in thermal/catalytic oxidation units
135	Destruction/removal efficiency (DRE) for dioxins/furans in thermal/catalytic oxidation units
136	Destruction/removal efficiency (DRE) for hydrocarbons and PICs in thermal/catalytic oxidation units
137	Effects of recycled scrub on melter and glass
138	Safe handling of melter feed during abnormal shutdown
139	Selection of initial baseline process configuration
140	Scrub purge recycle/blowdown ratio
141	Melter/Off-gas System Stream Factor
144	Melter feed rate basis
145	Composition of air
146	Acid scrub make-up composition
147	Caustic scrub make-up composition
148	NOxidizer fuel selection and fuel composition
149	NOxidizer reduction chamber air requirement
150	Selection of quench and scrub components in offgas treatment system
151	Chemicals included in mass balance calculations
152	Film cooler outlet temperature
153	Pressure control air requirement
155	Slagging/plugging control at melter outlet
157	Control of foaming in melter
158	Speciation and partitioning of sulfate in melter
159	Feed evaporation requirement

DBE ID	DBE Name
160	HgCl ₂ scrubbing efficiency in acid quench and venturi scrubber
162	SBW vitrification process is operationally safe against identified safety concerns
163	Fate of chloride and fluoride in the melter
164	Should Sr-90 be extracted from scrubber blowdown
165	Wash flow for HEME
166	Operating temperature in HEME
167	Soluble species in offgas
168	Operating temperature in demister
169	Caustic scrub purge rate
170	Scrub cooler outlet temperatures
171	Removal of particulate matter in the GAC beds
172	Removal efficiency for iodine in GAC
173	Conditioning of scrub purge prior to radionuclides extraction
174	NOxidizer reduction chamber chemistry using propane
175	Treatment and disposal path for spent Cs-137 IX resin

ATTACHMENT 2

MASS BALANCES FROM RSM-1, EV-16, RSM-2

The information in this attachment was prepared for the paper "Sodium Bearing Waste Vitrification Test Results", N.R. Soelberg et al, presented at the IT3'02 Conference, May 13-17, 2002, New Orleans, LA. It summarizes the mass balance/partitioning data obtained from the RSM-1, EV-16, and RSM-2 test series, described in the body of this report. The tables below (labeled "Table 10, Table 11, and Table 12", as in the referenced paper) were prepared using data supplied from several sources. The data used from each test, and their sources, are indicated in the table below:

TEST	DATA USED	SOURCE	DATA DESCRIPTION
RSM-1	Appendix C	Ref. 11 of main report	Additive data for SBW feed simulant
"	Table 4.3	"	SBW simulant composition
"	Spreadsheet "PapTbl.xls"	R. W. Goles (PNNL)	Glass composition data
EV-16	Table 3.1	Ref. 5 of main report	SBW simulant composition
"	Table 5.12	"	Off-gas flow rates
"	Table 5.16 Table 5.17 Table 5.18 Table 5.19	"	Off-gas composition data
"	Table 6.2	"	Feed species mass partitioning data in glass and offgas treatment streams
"	Table 6.3	"	Feed species mass partitioning data in offgas sample streams
RSM-2	Spreadsheet "RSM-2 Process Rates.xls"	R. W. Goles	Melter feed and glass production rates
"	Spreadsheet "GlsComp.xls"	R.J. Kirkham (INEEL)	Glass composition data
"	Spreadsheet "RSM feed Data.xls"	R.J. Kirkham	SBW simulant composition
"	Spreadsheet "rsms spread sheet scrubber data.xls"	R.J. Kirkham	Feed species mass partitioning data in offgas scrubber streams

The values in the tables below were generated in a single spreadsheet, "MB Calcs.xls". Due to differences in assumptions and interpretation of uncertainties in tabulated test data, the mass balance numbers in the tables here differ in many cases from those in the published reports referenced in the body of the current report. However, while absolute values may differ, data trends should be similar.

In the column headings of the tables "OG smpl" indicates data obtained from offgas sampling (as opposed to scrub system liquid/solid streams); "% in scrub", "% as gas" and "% on TF" indicate the fractions of feed captured in scrub streams, on total filter samples, and in gas sample impingers, respectively; "MB%" = overall mass balance closure. In Table 10 the MB% value is the sum of the first three column values, and in Tables 11 and 12 it is the ["% in glass"] value plus the greater of ["% in scrub"] and ["% on TF" + "% as gas"] values. Blanks in the tables indicate values for which data was not supplied in the data sources.

Table 10. Feed species mass partitioning for test RSM-1

	RSM-1 (B)				RSM-1 (D)				RSM-1 (E)			
	% in glass	% on TF	% as gas	MB%	% in glass	% on TF	% as gas	MB%	% in glass	% on TF	% as gas	MB%
Al	95			95	94			94	94			94
B	100	2.8	0.079	103	99	3.3	0.000	102	86	5.0	0.000	91
Ba	287	2.7	0.000	290	314	4.8	0.000	319	278	0.7	0.000	279
Ca	80	0.8	0.081	81	84	0.9	0.015	85	75	1.3	0.075	77
Cd	86	2.9	0.000	89	92	3.6	0.000	96	61	9.1	0.000	70
Ce	0.000			0.000	0.000			0.000	0.000			0.000
Cl												
Cr	184	5.0	0.000	189	240	5.3	0.000	245	215	7.7	0.000	222
Cs	54	20.0	0.000	74	55	20.0	0.000	75	54	33	0.000	88
Cu	50			50	40			40	56	6.7	0.000	63
F												
Fe	107	0.050	0.032	107	104	0.1	0.008	104	90	0.1	0.01234	90
Gd												
K	106	11	0.000	117	89	11	0.000	100	86	14	0.000	101
Li	99	2.3	0.000	102	97	2.6	0.000	100	85	3.3	0.000	88
Mg	178			178	147			147	143			143
Mn	123	0.3	0.000	124	111	0.2	0.000	111	105	0.3	0.000	105
Mo	0	0.4	0.000	0	0	1.1	0.000	1	0	0.8	0.000	1
Na	96	5.0	0.263	101	95	4.2	0.181	99	93	5.0	0.263	98
Ni												
NOx												
P	0	6.3	0.000	6					20	8.3	0.000	28
Pb	123			123	115			115	112	2.0	0.000	114
Ru	782	13	0.000	795	688	10	0.000	698	580	17	0.000	596
S	62	13	0.000	75	57	11	0.000	68	60	25	0.000	85
Si	100			100	101			101	89			89
Sr	1045	2.7	0.000	1048	474	1.9	0.000	476	328	2.5	0.000	331
Zn	83			83	82			82	84	0.6	0.115	85
Zr	15301			15301	3114			3114	3745			3745

Table 11. Feed species mass partitioning for test EV16-1

	EV16-1				
	% in glass	% in scrub	% on TF	% as gas	MB%
	82	0.1	0.000		82
	94	2.5	4.3		98
	84	0.6	0.000		85
	70	0.4	3.0		73
	98	32		1.7	131
	78	3.4	0.000		81
	33	20	0.000		53
	122	6.1	0.000		129
	301	24		0.02	325
	86	0.1	0.048		86
	91	0.2	0.000		91
	71	13	6.4		84
	80	2.0	0.000		82
	76	0.6	0.3		77
	3395	220.4	0.000		3615
	79	4.6	4.4		84
	118	2.9	1.7		121
	0	1.0		84	84
	80	1.3	1.2		82
	51	5.5	0.000		56
	53	11		1.4	64
	80	0.1			80
	113	2.2	0.000		115
	92	6.6	1.2		98

Table 12. Feed species mass partitioning for test RSM-2

	RSM-2 (2)						RSM-2 (3)						RSM-2 (4)					
	% in glass	% in scrub	% on TF	% as gas	MB%		% in glass	% in scrub	% on TF	% as gas	MB%		% in glass	% in scrub	% on TF	% as gas	MB%	
Al	84	0.8	0.3	0.000	85		77	0.1	0.2	0.000	77		77	0.4	NM*	NM	78	
B	83	4.6	3.8	0.000	87		89	2.9	0.7	0.3	92		89	4.7	NM	NM	94	
Ba	359	7.4	1.3	0.13	366		253	4.4	6.6	0.040	260		253	3	NM	NM	256	
C		1.7		132.8	133		0	3.0		0.000	0		0	3.1	NM	NM	3	
Ca	68	1.3	0.4	0.004	69		75	0.2	0.1	0.000	75		75	0.6	NM	NM	76	
Cd	62	7.4	1.9	0.000	69		54	3.5	1.0	0.000	57		54	8.6	NM	NM	62	
Ce	0	0.0			0		109				109		109	0.00	NM	NM	109	
Cl	83	19.9		13.1	103		80	6.6		0.000	86		80	5.4	NM	NM	85	
Co	0	0.0			0		0				0			0.00	NM	NM	0.00	
Cr	299	6.6	2.5	0.000	305		269	3.2	0.6	0.000	272		269	4.7	NM	NM	273	
Cs	80	9.9	11.1	0.003	91		95	5.7	2.0	0.003	101		95	8.7	NM	NM	104	
Cu	101	10.9	1.5	0.002	112		89	15.2	0.5	0.000	104		89	11.8	NM	NM	100	
F		51.9		123.4	123			43.7		334.1	334			107.7	NM	NM	108	
Fe	116	0.6	0.1	0.000	117		115	0.0	0.1	0.0	115		115	0.3	NM	NM	115	
Hg	0	23.1	0.68	71.5	72		0	16.1	0.50	165.6	166			19.2	NM	NM	19	
I		11.1		106.5	107		0	28.8		0.000	0			32.2	NM	NM	32	
K	104	7.8	7.3	0.000	112		100	3.5	1.2	0.000	103		100	5.6	NM	NM	105	
Li	96	3.4	2.2	0.000	99		99	1.1	0.3	0.000	100		99	2.1	NM	NM	101	
Mg	114	1.9	0.0	0.008	116		115	0.3	0.1	0.000	115		115	0.9	NM	NM	116	
Mn	54	0.7	0.3	0.000	55		51	0.2	0.1	0.000	51		51	0.4	NM	NM	52	
Mo	135	8.3	3.5	0.007	143		143	4.9	0.6	0.000	148		143	7.3	NM	NM	150	
Na	90	5.5	6.0	0.000	96		102	2.3	1.0	0.000	104		102	3.8	NM	NM	105	
Ni	494	11	0.1	0.000	504		251	16	0.1	0.000	268		251	12	NM	NM	263	
NOx				98	98		0			77	77			0.41	NM	NM	0.4	
P	114	2.5	0.5	0.000	117		110	0.8	0.1	0.000	111		110	1.6	NM	NM	111	
Pb	102	2.4	0.7	0.000	104		104	1.1	0.2	0.000	105		104	2.4	NM	NM	106	
Ru	756	9.8	4.9	0.000	765		92	8.6	2.9	0.000	101		92	13.0	NM	NM	105	
S	104	10.4	6.7	0.432	114		105	5.0	2.0	0.452	110		105	8.9	NM	NM	114	
Si	101	0.2			102		100	0.0			100		100	0.1	NM	NM	100	
Sr	100	0.000			100		105				105		105	0.0	NM	NM	105	
Ti	5367	12.8	0.0	0.000	5380		4697	2.6			4699		4697	6.5	NM	NM	4703	
Zn	61	2.5	1.4	0.104	63		66	1.3	0.6	0.073	68		66	1.2	NM	NM	68	
Zr	93	0.018	0.013	0.000	93		110	0.0	0.004	0.000	110		110	0.0	NM	NM	110	
V	87	1.4	0.2	0.000	88		107	0.2	0.1	0.000	107		107	0.6	NM	NM	107	

Table 12. (cont'd)

	RSM-2 (5)						RSM-2 (6)					
			OG smpl		MB%				OG smpl		MB%	
	% in glass	% in scrub	% on TF	% as gas			% in glass	% in scrub	% on TF	% as gas		
Al	84	NM	0.2	0.000	85	83	0.9	1.7	0.059	85		
B	83	NM	2.2	0.009	85	80	6.0	3.0	2.5	86		
Ba	359	NM	5.6	0.099	365	300	7.0	7.6	0.0228	308		
C	0	NM		0.000	0				0.000	0		
Ca	68	NM	0.3	0.001	68	68	1.3	1.2	0.0659	69		
Cd	62	NM	1.6	0.000	64	52	6.2	1.8	0.16	58		
Ce	0	NM			0	99				99		
Cl	83	NM		0.000	83	54	46.8		0.000	100		
Co	0	NM			0	0				0		
Cr	299	NM	0.8	0.000	300	325	6.1	2.5	0.43	331		
Cs	80	NM	3.1	0.003	83	87	8.6	2.3	0.38	95		
Cu	101	NM	0.5	0.000	102	92	32.5	1.3	0.12	125		
F		NM					52.3		14	14		
Fe	116	NM	0.071	0.000	116	113	0.5	1.3	0.0426	114		
Hg	0	NM	0.047	51	51	0	46.1	0.8	15	46		
I	0	NM		0.000	0		20.2		0.000	0		
K	104	NM	1.9	0.000	106	111	6.0	1.5	0.069	117		
Li	96	NM	0.8	0.000	96	98	2.8	1.4	0.13	100		
Mg	114	NM	0.1	0.004	114	108	1.8	1.2	0.11	109		
Mn	54	NM	0.3	0.000	54	92	1.2	2.8	0.069	95		
Mo	135	NM	0.8	0.000	136	160	6.0	0.9	0.090	166		
Na	90	NM	1.9	0.000	92	96	4.3	1.8	0.050	101		
Ni	494	NM	0.1	0.000	494	486	34.3	0.28	0.010	520		
NOx		NM					3.7		95.19	95		
P	114	NM	0.2	0.000	114	102	1.8	1.0	0.000	104		
Pb	102	NM	0.5	0.000	102	94	2.3	2.6	0.13	97		
Ru	756	NM	3.0	0.000	759	49	33.9	1.7	0.000	83		
S	104	NM			104	99	8.3	2.8	2.7	108		
Si	101	NM			101	100	0.1			100		
Sr	100	NM			100	96				96		
Ti	5367	NM	0.000	0.001	5367	5320	11.6	0.0	0.001	5332		
Zn	61	NM	0.7	0.13	62	64	2.2	2.6	0.72	67		
Zr	93	NM	0.002	0.000	93	130	0.0	0.5	0.000	130		
V	87	NM	0.1	0.000	87	101	0.6	0.4	0.013	102		

ATTACHMENT 3



INTEROFFICE MEMORANDUM

Date: November 5, 2001

To: T. T. Nichols MS 5218 6-9173

From: C. M. Barnes MS 3625 6-0864

Subject: EVALUATION OF SBW VITRIFICATION PROCESS ALTERNATIVES

Reference: D. Taylor, C. Barnes, L. Lauerhass, *INEEL SBW Vitrification Process*, INEEL/EXT-01-01139, September, 2001.

The mass balance for SBW vitrification will provide the basis for the conceptual design. A baseline flow scheme was defined in a value-engineering session held in mid FY 2001. During the value-engineering session, some variations to the baseline flowsheet were suggested but not evaluated. Also in the value-engineering session, some unit operations in the baseline flow scheme were defined more by function than equipment type, leaving room for later evaluations to provide additional flowsheet detail. FY 2001 end-of-year reports of vitrification testing and development also contain several recommendations for changes to the baseline flowsheet. This letter contains the results of a review of 33 process variations of the baseline, in order to determine (1) if test data is needed to evaluate the alternative (2) whether modeling can be of benefit evaluating the alternatives, and (3) for those that can be evaluated with presently available data and models, which alternatives would have the greatest impact on the mass balance.

A list of process these 33 alternative schemes is given in Table 1.

The process alternatives were initially screened relative to the questions

1. What data are needed to evaluate differences between the alternative and the baseline, and are these data available?
2. Can the Visual Basis/Excel mass balance model be used to evaluate differences between the alternative and the baseline?
3. Can the ASPEN PLUS mass balance model be used to evaluate differences between the alternative and the baseline?

Following this initial screening, the Visual Basic/Excel simulator was used to generate mass balances for those alternatives for which it was deemed appropriate. Results from the mass balances such as the glass rate, secondary waste compositions and rates, and internal stream flowrates were used to evaluate the alternatives.

The Technical Baseline Database (TBDB) is a repository of the baseline flow scheme, mass balance and basis information for the flow scheme and mass balance. The TBDB, documented as of August 2001 in the above reference, is a living document that will be updated as results from additional testing are received and additional modeling is performed. Results from this evaluation, once appropriately reviewed, will be incorporated into the TBDB.

Table 1. Process Alternatives
Baseline

Alternative

No SBW evaporation	1. SBW partial evaporation
	2. SBW total evaporation
	3. SBW denitration
	4. SBW absorption on silica gel
No sulfate removal from feed	5. Barium sulfate precipitation
No SBW feed blending	6. Feed blending from multiple Tank Farm tanks
Liquid sugar reductant	7. Granular sugar feed
	8. Alternative reductant
Glass frit	9. Multiple glass forming components
Film cooler	10. Alternative melter off-gas cooling
Spray quench tower*	11. Ejector venturi scrubber
Venturi scrubber*	12. Submerged bed scrubber
	13. Steam atomized scrubber
HEME*	14. Wet electrostatic precipitator
Single Scrub Tank	15. Cascaded acid scrub with multiple tanks
No HEME wash water tank	16. HEME wash water tank
Standard HEPA filters*	17. Washable HEPA filters
No Hg removal from scrub, all Hg removed by GAC	18. Hg removal from scrub by electrolytic reduction
	19. Hg removal from scrub by sulfide precipitation
	20. Hg removal from scrub by ion exchange (enhancements to Hg oxidation in the off-gas and Hg scrubbing from the off-gas would be considered for any of the above Hg removal schemes)
Split of acid scrub blowdown between recycle and secondary waste	21. Total recycle of scrub purge (may require Hg removal or other treatment)
Solids settling in acid scrub tank, filter on purge upstream of neutralization	22. No recycle of scrub purge
	23. No settling designed into scrub tank, filtration on total scrub flow (will require additional tank or tanks)
	24. Filter downstream of neutralization
Cs removal by IX from acid scrub purge	25. No treatment of acid scrub purge
	26. Removal of both Cs and Sr from scrub
	27. Removal of Cs and Sr downstream of combining acid and caustic scrub purge streams
Neutralization of acid scrub by caustic	28. Neutralization with other neutralizers (to reduce Na concentration in IX feed and to reduce volume of grouted waste)
Cs IX using Ionsiv IE-95	29. Cs IX using other sorbents
Caustic quench/scrub in single unit operation	30. Partial water quench only, no scrub
	31. Quench and scrub in separate steps
GAC bed upstream of final HEPAs	32. GAC bed downstream of final HEPAs
NO _x abatement using a noxidizer	33. NO _x abatement using SCR

* Alternatives for the in-cell off-gas solids removal operations may be evaluated as grouped combinations rather than single unit operation replacements

CONCLUSIONS

1. Pretreatment schemes have the potential for large savings, hundreds of millions of dollars, in Idaho Waste Vitrification Facility (IWFV) costs. These pretreatment schemes include
 - a. Barium sulfate precipitation followed by partial SBW evaporation with crystallization and separation of nitrate precipitates.
 - b. Barium sulfate precipitation without evaporation
 - c. SBW absorption on silica gel followed by evaporation to dryness
 - d. Partial evaporation of SBW, resulting in a slurry feed.

The magnitude of the potential savings estimated for these schemes justifies funding experiments to better quantify the expected improvements in such parameters as glass waste loading, frit formulation, melt rate, and melter off-gas rate. This additional test data would then be used in an evaluation that would more thoroughly select the recommended pretreatment process from the above four alternatives.

2. Two alternatives, replacing the film cooler with an evaporative cooler or replacing the noxidizer with an SCR reactor, have the potential for significant reductions in off-gas flowrates. If feed pretreatment is not incorporated into the baseline flowsheet, testing of these alternatives is recommended.
3. Five schemes (15, 18, 19, 21 and 22 in Table 1) should be evaluated using an ASPEN Plus model. Scrub composition data from RSM-2 and, if available, Clemson-2 should also be used in these evaluations.
4. Four additional changes to the baseline are recommended – replacing the sugar solution feed with granular sugar, blending SBW feed, changing the melter plenum temperature to 500°C and removing strontium from the acid scrub purge.
5. Unless new data is obtained that would suggest otherwise, no further evaluation is recommended of total denitration (scheme 3b in Table 1), chilling the air feed to the film cooler (10a), Hg removal by ion exchange (20), no treatment of acid scrub purge (25) or GAC downstream of final HEPA filter (32).

INITIAL SCREENING

Melter Feed Evaporation (Alternatives 1-4)

Alternatives 1-4 relate to partially or totally evaporating the feed to the melter. The four alternatives listed are representative of at least twenty permutations or variations of evaporation schemes, in which the variables include (a) the number of streams combined prior to the evaporation (out of the set: SBW, scrub recycle, reductant, one or more of the glass forming components), (b) degree of denitration (none, partial, total) (c) type of denitration (thermal, organic reductant, electrolytic, rotary kiln, fluidized bed), (d) number of process steps (separate denitration or combined with evaporation, removal and separate treatment of precipitated solids, etc.) (e) degree of evaporation, and (f) type of evaporator.

Test data and previous evaluations related to SBW evaporation and denitration schemes are contained in the following reports:

- (1) D. S. Wendt, D. R. Haefner, "Wet Vs. Dry SBW Vitrification Feed Evaluation," EDF-2860, October 11, 2001.
- (2) R. J. Kirkham, A. K. Herbst, "Suitability of Silica Gel to Process INEEL Sodium Bearing Waste," September, 2000.
- (3) R. J. Kirkham, "Sodium Bearing Waste Solidification by Evaporation," RJK-6-98, September 29, 1998.
- (4) J. A. McCray, "Report on Freeze Crystallization and Evaporation/Precipitation Testing for Sodium-Bearing Waste Treatment," JAM-11-98, September 30, 1994.
- (5) J. A. McCray, J. Pao, "High Temperature Evaporation/Precipitation Tracer Study Results," JAM-4-95/JHP-3-95, June 30, 1995.
- (6) E. P. Wagner, "Evaluation of Evaporation/Crystallization as Treatment for Sodium Bearing Liquid Waste," EPW-04-94, November 7, 1994.
- (7) R. D. Adams, "Evaporation/Precipitation Process for SBW Treatment Planning Estimate," RDA-03-94, October 27, 1994.
- (8) J. Pao, "Laboratory Simulation of Liquid Waste Evaporation," JHP-05-00, June 27, 2000.
- (9) D. W. Marshall, J. Pao, "FY-98 Idaho Nuclear Technology and Engineering Center Low Activity Waste Preconditioning Development Program Status Report," DWM-02-98/JHP-01-98, September 24, 1998.
- (10) J. Pao, "Comparison of High Temperature and Organic-Reductant Thermal Denitration of INTEC Low Activity Waste in a Bench Scale Rotary Reactor," JHP-04-98, December 24, 1998.
- (11) D. W. Marshall, "Optional Denitration Processes for Conditioning Low-Activity Wastes at the Idaho Nuclear Technology & Engineering Center," DWM-05-98, December, 1998.
- (12) D. T. Hobbs, *Electrolytic Treatment of ICCP Sodium-Bearing Waste Simulant*, WSRC-RP-94-1300, December, 1994.
- (13) D. D. Siemer, SAIC/Studsvik Calcination Test: Conclusions, DDS-08-00, August 29, 2000.
- (14) Tank Focus Area, "Technical Review of the Applicability of the Studsvik, Inc. THORsm Process to INEEL SBW," September, 2000.
- (15) W. H. Landman, "Solidification of SBW for EIS Supporting Studies," Conference Note, September 29, 1998.

Benefits from feed evaporation include an increase in the glass melt rate, a reduction in the melter heat requirement and a reduction in the off-gas flowrate. Wendt and Haefner¹ estimate a 50% increase in the glass rate for a dry feed compared to the baseline SBW composition. The increase in the glass rate would allow either for a 33% reduction in the processing schedule or, if the schedule were kept the same, a lower feed rate resulting in capital savings due to a smaller melter and smaller off-gas equipment. If the feed rate were kept the same as the baseline, the melter off-gas flowrate would be reduced by more than 60%, with corresponding reductions in downstream flowrates. The heat required to evaporate and superheat the water in SBW to the melter plenum temperature of 650°C is approximately 330 kW, more than twice as great as the estimated melter electrode power requirement of 160 kW.^a Wendt and Haefner¹ also estimate that the dry feed scheme equipment costs are about \$1 million less than the baseline wet feed, out of a total plant equipment cost of about \$30 million, due to savings in the feed system costs.

The major concerns or risks of feed evaporation relate to the transport, storage, handling and mixing of the concentrated SBW, either as sludge or "dry" solids. At a high degree of concentration, the concentrated SBW has been observed to "set up" upon cooling.³ If evaporated with the alkaline glass

^a Heat to evaporate and superheat waste is based on the baseline mass balance feed rate of 100 gallons per hour. The melter electrode power is taken from the SBW Vitrification Feasibility Study, which is based on a feed rate of 104 gal/hr.

forming components (GFCs) or a reductant, sufficient neutralization may occur to cause the precipitation of gelatinous aluminum hydroxide compounds.

The above benefits, as well as the amount of information currently available, justify leaving at least one feed evaporation scheme in the set to be further evaluated. In the following discussion, a rationale is developed for defining which evaporation schemes to retain and which to eliminate from further consideration.

As a result of road mapping efforts for SBW vitrification in late FY-00, the Tank Focus Area (TFA) sent an independent review team to the Studsvik treatment facility in Irwin TN to evaluate the applicability of the Studsvik process for evaporation and denitration of SBW. In their report,¹⁴ the TFA review team stated that the dry product that would be produced by the Studsvik process may be suitable as an interim storage waste and dry feed for the vitrification facility. However, they recommended that DOE-ID not pursue further steam reforming initiatives for treating SBW. The rationale for eliminating the Studsvik steam reforming process applies equally to other total denitration methods, including calcination and rotary kiln denitration. All of these processes involve large equipment operating at relatively high temperatures ($\geq 500^{\circ}\text{C}$). The cost of the denitration equipment does not justify the relatively small reduction in mass feed rate to the melter.^b In addition, denitration in a calciner or rotary kiln requires the addition of significant quantities of other materials such as aluminum nitrate that would add to the melter feed mass flowrate and the glass rate.

Proof of principle tests were performed¹² on a low temperature denitration method, electrolysis, using an SBW simulant. However, this alternative is rejected due to (a) low denitration efficiencies (55-68%), (b) the explosion hazard of producing hydrogen gas at the cathode, (c) the potential for metals and metal hydroxide films plating out on the electrodes, (d) additional processing steps required due to the ammonia gas produced, and (e) the lack of data available for this process.

In the present baseline mass balance, the acid scrub recycle rate is determined by the buildup of chloride in the scrub. Total solids (dissolved plus undissolved solids) in the scrub are much lower than in SBW, hence the water content is higher (95+% compared to ~65% for SBW). The scrub recycle rate is only about 5-10% of the fresh SBW rate, hence separate evaporation of the scrub would result in only small benefits that would likely not be justified by added equipment costs. However, because of the high water content of the recycle relative to SBW, it makes sense to combine the scrub with SBW prior to evaporation. If the evaporation were to the point of dryness, combining the scrub would eliminate any liquid feed to the melter.

The question of whether to combine other melter feeds, i.e., the reductant or some or all of the GFCs prior to evaporation is less easily answered. The benefit of combining all melter feeds prior to evaporation relates to mixing. If GFC and sugar are combined with the liquid feeds, the final concentrate could be expected to be more homogeneous than if added after evaporation of the SBW. Adding the alkaline GFCs (Li_2O , CaO and Na_2O) to the liquid SBW would result in a partial neutralization of SBW, which would reduce the amount of nitrate removed (as nitric acid) in the evaporation. Adding sugar prior to evaporation has the advantage of destroying, at evaporation temperatures, all the nitric acid. If sugar is not added, the nitric acid removed in evaporation would be recovered as concentrated nitric acid as LET&D bottoms. If no use can be found for this acid, it would need to be processed into a separate waste form. Destroying the nitric acid by the addition of sugar prior

^b Nitrate associated with nitric acid in the feed can be removed by evaporation or reduction with sugar. Nitrate in excess of that associated with acid amounts to about 15-20 wt % of SBW, or 10-13 wt % of the total melter feed.

to evaporation would eliminate this waste stream. However, it would tie the evaporation more closely to vitrification off-gas system, as noncondensibles from the evaporator condenser would need to be added into the melter off-gas upstream of the noxidizer.

The following table summarizes the screening of the remaining potentially viable SBW evaporation schemes. It should be noted that the primary reason several of the schemes were screened out is lack of data. One partial-evaporation scheme and one total-evaporation scheme are retained. Others that are screened out at this time may be worth reconsidering at a later time if additional test data is obtained.

Table 2. Screening decisions for SBW evaporation schemes

<u>Scheme</u>		
<u>Feed</u>	<u>Degree of Evaporation</u>	<u>Screening Decision and Basis</u>
SBW & Scrub Recycle	Partial	Retain based on results of McCray's evaporation tests and evaluations ⁴⁻⁷
SBW & Scrub Recycle	Total	Reject based on lack of data and potential problems with properties of solid product
SBW, Recycle & GFC	Partial	Reject as this option reduces the amount of acid removed by evaporation
SBW, Recycle & GFC	Total	Retain based on Kirkham's SBW silica gel absorption tests
SBW, Recycle & Sugar	Partial	Reject based on lack of data and potential problems with properties of solid
SBW, Recycle & Sugar	Total	Reject based on lack of data and potential problems with properties of solid
SBW, Recycle, Sugar & GFC	Partial	Reject based on lack of data and potential problems with properties of solid
SBW, Recycle, Sugar & GFC	Total	Reject based on lack of data and potential problems with properties of solid

A final question needing an answer in order to define the retained partial evaporation scheme is, "To what extent should the feed be evaporated?"

McCray⁴ evaporated an SBW simulant in increments of approximately 20% by volume. At the end of each incremental evaporation, the solution was cooled, and if precipitates formed, they were collected. The simulant McCray used had concentrations of 0.53 M Al^{+3} , 1.17 M Na^{+} , 5.5 M NO_3^{-} , as well as lower concentrations for 20 other species. No precipitation was observed until his third increment. Hence the onset of precipitation was somewhere between 37% and 53% volume reduction. The concentration of Al^{+3} in the current "total" SBW composition is 4% higher than McCray's simulant, Na^{+} is 30% higher, and nitrate 8% higher. WM-180 has an Al^{+3} concentration 25% higher than McCray's and Na^{+} concentration 76% higher. Comparing the concentrations of the present waste streams to McCray's results suggests that a volume reduction of 10-25% could be achieved before the onset of precipitation. Test results from Marshall and Pao^c lead to a similar conclusion.

^c D. Marshal and J. Pao evaporated an SBW LAW simulant, essentially diluted SBW. They achieved about 70% volume reduction before precipitates formed. However, when the results are adjusted to present SBW concentrations, the degree of concentration achievable appears to be closer to 10-15%.

Evaporation only to the point of precipitation would avoid problems due to solids in the evaporator equipment for the process scheme of evaporation of SBW and scrub recycle only. However, if the reductant and/or GFC were added, much of the nitric acid would decompose or react, the solution pH would increase, and precipitation would be likely. Thus for the scheme in which one or both of these other components is added prior to evaporation, it is likely that even less concentration than the above estimate could be achieved prior to precipitation.

The melter feed in the baseline mass balance contains approximately 15 wt % undissolved solids and 50 wt % water.^d The process basis for the DWPF SME is concentration to 50 wt % total solids.⁴ Nearly all the solids in the DWPF feed are undissolved, since Savannah River sludge has been washed prior to entering the DWPF process. Ideally the degree of evaporation to a slurry would depend on rheological property limits for transferring the slurry from the evaporator through melter feed tanks and into the melter. One DWPF mass balance^e shows the following limits for SME concentrate:

	Minimum	Maximum
Yield stress, dynes/cm ²	25	150
Viscosity, cp	10	49
Density, g/cm ³	1.33	1.45
Solids content, wt %	40	50

A more recent report^f of DWPF operation shows the density of SME concentrate in the range 1.39-1.47 g/cm³. Feed solids content of up to 59% have also been reported.^g Until these target properties can be better defined, the DWPF maximum limits provide a reasonable starting target concentration. Based on a 50% total solids content, the degree of concentration would be about 50-60%.

Process simulation can help determine the benefits of partial evaporation, however, testing would be required to validate major assumptions in the mass balance. Testing would be needed to more clearly define the degree of evaporation and to determine the glass melt rate for concentrated feeds.

Barium Sulfate Precipitation (Alternative 5)

Recommendations of the waste loading of SBW (as oxide) in glass have decreased from 36% (April, 1999)^h to 35% (July, 1999)ⁱ to 30% (2001)^j to 20% (2001)^k as more and more data is collected. Data has shown that the waste loading is limited by the sulfate content of the SBW. Removal of sulfate from the feed thus has the potential for greatly reducing the glass volume, which in turn would reduce storage, shipping and other costs. Removal of sulfate from the feed would also benefit the off-gas and scrub system by reducing the amount of SO₂ in the off-gas.

^d Solids from the insoluble frit components in the total melter feed vary between 13% and 21% for the four mass balance cases in Reference 2, while the total water from SBW, recycle and sugar amounts to 48-50% depending on feed case.

^e Basic Data Report, DWPF Sludge Plant, DPSP 80-1033, Rev. 9, September, 1982.

^f J. E. Occhipinti, J. T. Carter, R. E. Edwards, R. S. Beck, D. C. Iverson, *DWPF Radioactive Operations - Year Two*, WSRC-MS-98-00108, 1998.

^g J. M. Perez, Jr. et. al, *High Level Waste Melter Study Report*, PNNL-13582, July, 2001.

^h D. K. Peeler, J. D. Vienna, *Waste Loading Estimates for INEEL HAW*, WSRC-RP-99-00349, April 22, 1999.

ⁱ J. D. Vienna, et.al, *Glass Formulation Development for INEEL Sodium-Bearing Waste*, PNNL-12234, July, 1999.

^j D. K. Peeler, T. B. Edwards, I. A. Reamer, R. J. Workman, J. D. Vienna, J. V. Crum, M. J. Schweiger, *Glass Formulation Development for INEEL Sodium-Bearing Waste (FY-2001 WM-180)*, WSRC-TR-2001-00295, September 21, 2001

^k J. D. Vienna, "SBW-22," e-mail to K. J. Perry, July 12, 2001.

Data regarding barium sulfate precipitation is found in the following reports:

- (16) S. K. Fiskum, D. E. Kurath, B. M. Rapko, "Development and Demonstration of a Sulfate Precipitation Process for Hanford Waste Tank 241-AN-107," PNWD-3050, BNFL-RPT-029, August, 2000.
- (17) C. M. Nitzel, "Separation of Barium Sulfate Precipitate from Liquid Sodium-Bearing Waste Simulant," August, 2001.
- (18) D. D. Siemer, unpublished results of barium sulfate precipitation tests, March 3, 2001, July 3, 2001, July 17, 2001

The above reports show that 60-90% of the sulfate in a waste can be removed by precipitation as barium sulfate. Adding barium to an alkaline waste such as Hanford's results in precipitation of barium carbonate as well as barium sulfate.¹⁶ However the acidity of SBW will prevent carbonate formation. Using test surrogates spiked with radionuclides Siemer¹⁸ showed that about 1-2% of the plutonium and 2-3% of strontium in the initial test sample was contained in the sulfate precipitate. The Hanford tests¹⁶ showed that the sulfate precipitate contained only 0.02% of the ¹³⁷Cs in the feed, less than 1% of the ⁶⁰Co, ¹⁵⁴Eu and ¹⁵⁵Eu, but 10% of the ⁹⁹Tc. Additional test data may be needed to determine the radionuclide concentrations in the precipitate to determine its disposal site. Nitzel¹⁷ demonstrated that the precipitate could be separated from the waste solution by either settling or filtration.

Sufficient data is available to develop an initial mass balance for a barium sulfate precipitation scheme. Because of the benefits of removing sulfate from the feed, this process alternative is retained for further evaluation.

SBW Feed Blending (Alternative 6)

Based on tank farm planning documents updated as recently as May 31, 2001, SBW from five different tanks would be blended in new tanks in 2010, providing a homogeneous liquid feed to the melter. A feasibility study¹ for the new tank farm is presently being completed. However, because of uncertainties regarding whether a new tank farm would be constructed, the SBW Vitrification report^m contained mass balances for three individual SBW tanks as well as the blended total SBW. At present, one tank (WM-180) is full and no changes to its composition are expected prior to either transfer to a new tank or processing in the vitrification facility. Waste presently in several other tanks, along with other waste that will be generated until 2005, will be evaporated and make up the remaining vitrification facility feed.

In September 2001, tank farm management plans were changed in order to fill a second tank in FY-02. According to these revised plans,ⁿ tank WM-189 will be filled in January, 2002 and WM-188 in June, 2006. Besides the SBW waste in these three tanks, about 100,000 gallons will be present in other tanks in 2010.

The SBW vitrification mass balances^m show that the total glass produced from 865,000 gallons in WM-180, WM-188 and WM-189 is 619 m³ and the total grout is 494 m³. The glass and grout volumes produced from 960,000 gallons of a blended waste are 618 m³ and 521 m³ respectively. Prorating the glass and grout volumes produced from the three tanks up to the total volume of SBW shows that

¹ K. Childs, et. al, *Idaho Waste Vitrification Facility Project Waste Collection Tank Facility Feasibility Study Report (draft)*, September 10, 2001.

^m D. Taylor, C. Barnes, L. Lauerhass, *INEEL SBW Vitrification Process*, INEEL/EXT-01-01139, September, 2001.

ⁿ C. B. Millet, Excel file PEMP 2002 HLLWE Case-shortfile2, September 26, 2001.

blending would result in about 68 m³ less of glass and 27 m³ less of grout. While these volume differences contain uncertainties because of uncertainties in projected waste compositions and other mass balance assumptions, they do suggest that blending offers significant savings.

A comparison of compositions of waste in WM-180, WM-188 and WM-189 is shown in Table 2, expressed as concentrations in a tank compared to the average. For a few species, concentrations in each of the 3 tanks exceed the average because concentrations in the remaining waste are less than the average.

Table 2. SBW Composition Variation
WM-180/ WM-188/ WM-189/
Average Average Average

H+	0.44	1.42	1.02
NO3	0.94	1.10	1.13
Al	1.24	1.06	1.16
B	0.70	1.26	1.50
Ca	0.96	1.17	1.39
Cr	0.75	1.35	1.44
Cs	0.58	1.81	1.17
Fe	1.04	1.22	1.22
Hg	0.58	1.46	0.68
K	1.25	0.96	1.29
Na	1.42	0.89	1.20
U	0.76	1.14	1.22
PO4	2.00	0.61	0.94
SO4	1.45	0.90	1.12
U-235	0.82	1.26	1.20
U-238	0.73	1.32	1.24
Np-237	0.23	1.85	1.11
Pu-238	0.97	1.27	1.05
Pu-239	1.20	1.22	0.87
Am-241	0.49	1.83	0.88
Sr-90	0.29	1.95	0.93
Tc-99	0.81	1.71	0.72
Cs-137	0.55	1.76	0.90

Test data is needed to more accurately determine the effects of composition on glass loading, the need for multiple glass frits, effects on scrub composition, melter separation factors, and other mass balance assumptions. However, based on the differences in waste volumes between processing the SBW tank waste separately or blended, and judgement that variations in feed composition of the magnitude shown in Table 2 will affect design requirements and costs for many equipment items of the vitrification process, it is recommended that until data is obtained that shows these effects do not more than offset any incremental costs associated with blending, blending be assumed for the baseline flowsheet. Because mass balances prepared for the SBW Vitrification^m report include cases both with and without blending, no additional mass balances are needed to evaluate this alternative.

Granular Sugar Feed (Alternative 7)

In the present mass balance, the reductant is fed as 67 wt % sugar solution. Heating is required to keep the viscosity of this syrupy solution sufficiently low to pump. The water present in the sugar solution amounts to about 8-10% of the total water in the melter feed, assuming no feed evaporation. Melter tests to date have used granular rather than dissolved sugar. If the feed is evaporated to dryness, a granulated sugar feed is recommended to keep the feed as flowable solids and eliminate reactions that may occur in feed storage in the liquid phase between sugar and nitrate in the waste. Cost differences between equipment to process liquid and granular sugar are expected to be small. However the benefit appears to be in the direction of granular sugar due to (a) eliminating the need to replace the sugar feed system when revamping the vitrification facility for calcine, (b) elimination of the need to heat the stored sugar, (c) reducing the amount of water fed to the melter, and (d) aligning the design basis more with melter tests. Hence it is recommended that granular sugar be assumed for the baseline flowsheet. This decision can be revisited during conceptual design as details of the sugar feed system are determined.

Alternative Reductants (Alternative 8)

The use of sugar as the reductant in the melter is based on test data contained in the following report:

- (19) J. A. McCray, D. L. Griffith, R. R. Kimmitt, D. D. Siemer, *Status of Melt Rate Testing and Reductant Selection for SBW Vitrification*, INEEL/EXT-01-01223, September, 2001.

The above report found that activated carbon, glycolic acid and corn starch did not meet the requirements set for the reductant, but sugar did. The report concluded that "sugar will likely be the preferred reductant." However, additional testing was recommended. At this time there is no basis for generating mass balances with alternative reductants. If these additional tests are performed, results from these additional tests could provide the basis for evaluating different reductants for the SBW vitrification process.

Glass Forming Components (Alternative 9)

The glass "frit" formulation in the baseline mass balance is based on test data contained in the following report:

- (20) D. K. Peeler, T. B. Edwards, I. A. Reamer, R. J. Workman, J. D. Vienna, J. V. Crum, M. J. Schweiger, *Glass Formulation Development for INEEL Sodium-Bearing Waste (FY-2001 WM-180)*, WSRC-TR-2001-00295, September 21, 2001.

The "frit" formulation planned for future tests contains nine components - B_2O_3 , CaO , Fe_2O_3 , Li_2O , MgO , Na_2O , SiO_2 , V_2O_5 , ZrO_2 .^o In SBW surrogate melter tests to date, glass-forming components (GFCs) have been used rather than a prepared glass frit. Different compositions consisting of oxides, hydroxides or carbonates of Si, B, Li, Fe, Ti, Ca and Ba were evaluated in determining the recommended composition. Melter tests planned for FY-02 will include fritted glass formers, and the results from these tests will provide a basis for selecting either frit or GFCs. Frit has been assumed for the baseline flowsheet to minimize the number of feed storage tanks and additional water present as hydrates in

^o J. D. Vienna, Excel spreadsheet "SBW-22 comps," attached to e-mail "SBW-22", J. D. Vienna to K. J. Perry, July 12, 2001.

GFCs. Apart from this difference in water content, mass balance simulations of these two alternatives would not show differences, hence will not be performed in this evaluation.

Alternative Melter Off-gas Cooling (Alternative 10)

The baseline design uses a film cooler to prevent particulate and molten solids in the melter off-gas from depositing on the walls of the off-gas piping. Air and/or steam are added in the film cooler to cool molten particulate in the melter off-gas. The air added in the film cooler adds a significant fraction to the total off-gas rate. Barnes^p calculated that a reduction in melter plenum temperature from 650°C to 500°C would result in a 20-30% reduction in all downstream flowrates. Peurrung^q calculated that replacing a film cooler with evaporative cooling would reduce the downstream flowrate for the Hanford low-level waste vitrification facility by 57%.

The melter plenum temperature is related to the feed composition, the feed rate, the melt rate and the cold cap coverage. For the DWPF melter, the vapor space above the cold cap was kept at a temperature between 650°C and 800°C to provide sufficient heat for evaporation and melting.^r Recent optimization resulting in dropping the minimum plenum temperature of the DWPF melter to 490°C.^s The target plenum temperature for FY-01 tests with SBW surrogate at the Clemson Environmental Technologies Laboratory^t was 600°C, and varied between 450°C for a low feed flowrate to about 650°C for a high feed flowrate. In the Clemson tests, the “high” feed rate was 87% greater than the “low” feed rate. Future design studies should include optimization of melter plenum temperature relative to melter and off-gas system costs, and additional melter tests should provide data to establish the minimum required plenum temperature.

Process variations to the baseline flowsheet include two alternatives. Changing the air to steam ratio or chilling the film cooler air could result in small reductions in the off-gas flowrate. A larger reduction could be achieved by replacing the film cooler with an evaporative cooler. The Studsvik process uses an evaporative cooler to cool off-gas from the reformer. Another evaporative cooler design, called the “transpiring wall reactor” is somewhat similar in design to a film cooler, except that water replaces the air or steam used to maintain a clean fluid boundary layer to protect walls of a reactor from solids deposition. The transpiring wall reactor^u was developed to prevent molten salts that precipitate in supercritical water oxidation reactors from depositing on the reactor wall.

Mass balances are presented in a later section of this report that better quantify potential benefits of these alternatives. However, test data would ultimately be required to demonstrate the feasibility of any evaporative cooler design and confirm its expected performance.

^p C. M. Barnes, “Mass Balance Sensitivity Analysis and Process Alternatives,” CMB-11-01, September 27, 2001

^q L. M. Peurrung, T. J. Deforest, J. R. Richards, “Process System Evaluation – Consolidated Letter Reports. Volume 1 – Alternatives for the Off-Gas Treatment System for the Low-Level Waste Vitrification Process,” PNNL-11056, March, 1996.

^r DWPF Melter Technology Manual, Section 5, Reference 1, Processing Facilities Basis, Glass Melting, DPSP-80-1033, Part 5, Item 255, June, 1984.

^s D. Whit, personal communication, Dec. 4, 2001.

^t K. J. Perry, R. R. Kimmitt, N. R. Soelberg, R. D. Tillotson, A. N. Olson, Test Results for SBW-FY01-PS-01 Vitrification Demonstration of Sodium-Bearing Waste Simulant Using WM-180 Surrogate, INEEL/EXT-01-01073, August, 2001.

^u B. L. Haroldsen, D. Y. Ariizumi, B. E. Mills, B. E. Brown, D. C. Rousar, Transpiring Wall Supercritical Water Oxidation Test Reactor Design Report, SAND96-8213, February, 1996; S. F. Rice, B. C. Wu, W. S. Winters, C. D. Robinson, Engineering Modeling of the Pine Bluff Arsenal Supercritical Water Oxidation Reactor, SAND2000-8656C, April 9, 2000.

Melter Off-gas Scrubbing (Alternatives 11-14, 17)

Alternative systems and equipment for scrubbing solids and acid gas from Idaho Waste Vitrification Facilities (IWVF) melter off-gas are discussed in the following report:

- (21) R. Wood, D. Tyson, B. Bonnema, C. Olsen, A. P. Pinto, D. Wendt, S. Reese, B. Raivo, Feasibility Study for the Idaho Waste Vitrification Facilities Off-gas Treatment for Sodium-Bearing Waste, INEEL/EXT-01-00995, September, 2001.

The baseline off-gas scrubbing components are expected to meet process functional and operational requirements that have been identified for the IWVF. At the present time there is considerable uncertainty in melter DFs, off-gas mercury speciation, particle size distribution (PSD) of particulate in the off-gas, and NO_x , SO_x and Hg reactions that occur prior to and within the early off-gas treatment steps. Optimization and evaluation of off-gas scrubbing components will require additional melter off-gas characterization data and data from tests using the alternative and baseline scrubbing components. Alternative systems could be required if future testing (a) shows baseline components do not perform as presently assumed, (b) melter off-gas characterization is different from what is presently assumed in ways that affect scrubbing requirements, or (c) alternative components are shown to offer significant cost or performance advantages.

Cascaded Acid Scrub (Alternative 15)

The baseline flowsheet contains a single acid scrub tank which collects liquid from the quench tower, venturi scrubber and HEME. The scrub tank acts as a settling tank to concentrate solids for recycle to the melter and return liquid relatively free of solids to the quench tower and venturi scrubber.

The SBW Vitrification Feasibility Study^v shows a cascaded scrub system which includes separate collection tanks for venturi scrub and quench tower liquids. Make-up is supplied to the venturi scrub collection tank, and overflow from the venturi scrub tank is sent to the quench tank. No solids separations is done in the collection tanks, but rather in a separate tank containing the purge from the quench tank.

Additional off-gas solids characterization data is needed to better evaluate the feasibility of separating solids in the scrub tank. If solids separation in the scrub tank is feasible, the single tank scheme should remove solids from the off-gas with greater efficiency than the cascaded system. This is because in the cascaded system, the amount of solids returned in the scrub and quench solutions to the quench tower and venturi scrubber is several times greater than that contained in the off-gas coming to these unit operations. A cascaded scrub system would be beneficial if the acid scrub components had requirements to remove soluble species from the off-gas. However, no requirement for removal of any soluble species has yet been identified for the acid scrub system.

The make-up to the acid scrub is water; the scrub is acidic only because of the absorption of NO_2 by water in the quench tower and venturi scrubber. The acidity of the scrub will influence whether particulate captured in the scrub will dissolve. Dissolution of certain radionuclide and hazardous species affects secondary waste compositions. Thus, if the scrub acidity is different from what has been assumed, it may affect which scheme is preferable.

^v J. J. Quigley, B. D. Raivo, S. O. Bates, S. M. Berry, D. N. Nishioka, P. J. Bunnell, Feasibility Study for Vitrification of Sodium-Bearing Waste, INEEL/EXT-2000-00952, September, 2000.

Until additional off-gas solids and scrub composition data is received, mass balances help in evaluating different scrub schemes. After data from the RSM-2 tests and the Clemson-2 tests is received, this decision can be reevaluated. This additional data plus the phase equilibrium capabilities of ASPEN Plus may indicate that ASPEN Plus simulations can be of benefit in evaluating the different scrub schemes.

Separate HEME Scrub Tank (Alternative 16)

The Feasibility Study for Vitrification of Calcine^w assumes an off-gas treatment scheme that includes a separate tank for receiving water from and recycling water to the HEME. The baseline flowsheet uses make-up water supplied from the make-up water header, along with condensate from cooling coils within the HEME or an external cooler to the HEME. Cost differences between these two schemes are expected to be very small. The rationale behind the baseline flowsheet was that at times the HEME liquid would contain significant solids and recycling these solids to the HEME could negatively impact its performance. Evaluation of this alternative should be deferred to future design studies, after additional test data provides better characterization of the HEME liquid solids and dissolved specie concentrations.

Mercury Removal from Scrub (Alternatives 18-20)

Based on the present mass balance assumptions, the concentration of mercury in the acid scrub will increase to an equilibrium level. For the average SBW feed, the mercury concentration in the scrub reaches 28 g/liter, for WM-189, the concentration reaches 53 g/liter.^x Mercury concentrations in the grouted scrub purge for these two cases are 390 mg/kg and 580 mg/kg respectively. Because the concentration of mercury puts the grout into the “high” mercury (greater than 260 mg/kg) category, land disposal would require obtaining an equivalency agreement with the disposal site and regulating agencies. Amalgamation is the required treatment process for high mercury waste; thus grouting would need to be demonstrated and accepted as producing equivalent results.^y While the concentration of mercury in the grouted waste puts the waste in the high Hg category, the fraction of mercury in the feed that is contained in the grout is small, only 1-4%, depending on the feed case. The bulk of the mercury is contained in the activated carbon, producing a high volume (54 m³) of a second high mercury waste. An equivalency agreement would also be needed to dispose of the spent carbon. Removal of mercury from the scrub provides two benefits – reducing the level of mercury in the grout to below 260 mg/kg, and reducing the quantity of spent carbon waste. These advantages need to be weighed against the costs of additional equipment to remove mercury from the scrub plus the disposal costs an additional waste.

Data defining mercury speciation in the off-gas is needed in order to better determine mercury scrub concentrations and to evaluate mercury removal methods. Data from RSM-2 may provide a basis for initial evaluations. Additional information that can be used in this evaluation is contained in the following:

- (22) J. A. DelDebbio, L. G. Olson, J. Pao, Final Report on Mercury Vapor/Liquid Equilibrium to Support Wet Scrubber Process Models for NWDF Upgrade, JAD-02-2000, June 19, 2000.

^w S. O. Bates, B. D. Raivo, J. J. Quigley, S. M. Berry, W. H. Landman, S. L. Palmer, T. M. Hipp, Feasibility Study for Vitrification of Calcine in the Idaho Waste Vitrification Facility, INEEL/EXT-01-00978, September, 2001.

^x The “equilibrium” level of mercury in mass balance off-gas streams is determined by assumed separation factors, not vapor-liquid equilibrium (VLE) data.

^y A. K. Herbst, R. J. Kirkham, S. J. Losinski, Secondary Waste Considerations for Vitrification of Sodium-Bearing Waste at the Idaho Nuclear Technology and Engineering Center FY-2001 Status Report, INEEL.EXT-01-01172, September, 2001.

- (23) D. W. Marshall, Mercury Removal from Adjusted 10-cm Pilot Plant Scrub Solutions by Electrochemical Reduction, DRM-02-00, May 25, 2000.
- (24) J. A. DelDebbio, L. G. Olson, Status Report on Wet Scrubbing for Mercury Removal from Simulated NWCF Off-Gas, JAD-01-00, April 19, 2000.
- (25) D. D. Siemer, The "Hows & Whys" of Electrodeposition of Mercury from NWCF Off-gas Scrub, Recent Experimentation, DDS-06-00, August 15, 2000.
- (26) D. D. Siemer, Reprise of "Scrubability of Mercury from NWCF Off-gas," DDS-05-00, August 8, 2000.
- (27) D. R. Marshall, Mercury Removal from Simulated INTEC Acid Waste Solutions by Electrochemical Reduction, DRM-01-99, September 29, 1999.
- (28) R. E. Schindler, Status Report on Development of an ASPEN Electrolyte NRTL Parameter Set for Process Simulation of INTEC Aqueous Wastes, INEEL/INT-2000-0025, January, 2000.
- (29) S. C. Ashworth, et. al, NWCF Mercury Removal Feasibility Study, INEEL/INT-2000-00539, September, 2000.

It is recommended that mercury removal evaluations be done using the ASPEN mass balance model, because of the importance of Hg vapor-liquid equilibria (VLE) in the results. The mass balance model used in the past does no (VLE) thermodynamic calculations, but relies on assumed (inputted) separation factors. Comparison of recent mass balances to VLE data in DelDebbio²² shows that the mass balance scrub Hg concentrations are low by factors of 1.7-5.1. The VLE data generated by DelDebbio²² can be used as a check of ASPEN NRTL parameters derived by Schindler.²⁸

The previous mercury testing and evaluations have concerned mercury removal from the NWCF scrub and off-gas,²²⁻²⁹ and have focused on scrubbing mercury from off-gas and removal of mercury from NWCF scrub by electrochemical reduction. The data in these studies will be of help "calibrating" an ASPEN mass balance model to then use in simulating the vitrification process. The only feasible method of removing mercury from NWCF scrub solution was determined to be by electrochemical reduction. However, the vitrification scrub solution is not expected to be nearly as acidic as NWCF scrub, hence mercury removal methods other than electrochemical reduction may also be feasible. Sulfide precipitation would require neutralizing the scrub. Removal by ion exchange may be feasible, but would generate larger quantities of waste than electrochemical reduction or sulfide precipitation. Therefore analysis by ASPEN modeling is recommended only for these later two alternatives.

Acid Scrub Blowdown Split between Recycle and Purge (Alternatives 21-22)

Total recycle of the acid scrub purge to the melter would eliminate the acid scrub purge and hence eliminate the equipment in the baseline flowsheet to neutralize and remove cesium from the acid scrub purge. Total recycle would also reduce the quantity of grouted waste. Zero recycle of scrub purge would reduce the quantity of glass and increase the quantity of grout. The grout waste classification would likely change, becoming a TRU mixed waste. With no recycle of scrub, the melter feed composition can be more easily determined and will likely be more homogeneous. Total recycle would likely require removal of mercury from the scrub. Mass balance simulations can provide an initial evaluation of these two alternative schemes. Some assumptions used to generate the mass balances may need to be validated by test data. Because VLE will have a strong impact on these mass balances, it is recommended that the ASPEN model be used.

Filtration of Scrub Solids (Alternatives 23-24)

In the baseline flowsheet, the acid scrub tank acts as a settling tank to minimize solids in the quench and venturi scrub feeds and concentrate solids in the scrub recycled to the melter. The purge acid scrub is filtered, neutralized and then passes through ion exchange columns to remove cesium. Scrub solids characterization data are needed to confirm the feasibility of separation by settling. If not feasible, an alternative solid/liquid separation system would be needed.

Questions have been raised regarding whether precipitation will occur when the scrub purge is neutralized. If precipitates do form, the filter on the scrub purge should be moved downstream of neutralization.

Relative to the mass balance, alternative schemes 23 and 24 result in negligible changes to waste product quantities and compositions and most flowrates through the process. Thus it is recommended that any evaluation of these two alternatives be deferred until additional data is obtained on (a) scrub solids and (b) acid scrub neutralization.

Treatment of Scrub Purge (Alternatives 25-29)

Several variations of the baseline flowsheet have been suggested that relate to treatment of the scrub purge. One alternative, suggested in order to simplify the process, is to eliminate the ion exchange columns. This change would eliminate both the ion exchange removal equipment and the spent cesium ion exchange waste. Alan Herbst has addressed this suggestion in his recent report on secondary Waste Considerations:^z

“If the cesium were left in the (grouted) waste, the waste form would be (Hanford) Category 3/Class C. The real answer to this issue is in the grouting process as to whether the process will be contact handled or remote handled. By removing the cesium, the radiation dose from a drum of grout is reduced from over 900 millirem per hour (mR/hr) to less than 1 millirem per hour. The trade off is then the requirement to add shielding to the grouting process for remote handling of 900 mR/hr as opposed to no shielding for contact handled. It is thought that the expense of an ion exchange system is less expensive than a remote-handled, shielded grout mixing system.”

For reasons stated by Herbst, primarily that the expected cost and complexity of a remote-handled grout mixing system exceeds the cost of the ion exchange system, it is recommended that no further evaluation be performed on the scheme which deletes the cesium ion exchange column.

In the same report,^z Herbst recommends combining the acid and caustic scrub purges, and then removing cesium and strontium from the combined stream. Removal of strontium reduces disposal costs by reducing the waste classification from Hanford as Category 3 to Hanford Category 1. Approximately 90% removal of the strontium would be required to meet Hanford Category 1 waste limits. The sorbent used for cesium removal in the baseline process, IONSIV IE-95, does remove strontium as well, although the present mass balance does show any strontium removal because insufficient data is available on the scrub solution composition. Because of expected savings in disposal costs by removing strontium, this alternative is recommended. Additional test data is needed to

^z A. K. Herbst, R. J. Kirkham, S. J. Losinski, Secondary Waste Considerations for Vitrification of Sodium-Bearing Waste at the Idaho Nuclear Technology and Engineering Center FY-2001 Status Report, INEEL.EXT-01-01172, September, 2001.

determine whether the existing cesium sorbent can achieve the required strontium removal efficiency, or if not, what sorbent should be used.

If a higher pH is required for efficient strontium removal, combining the acid and caustic scrub purges may be of benefit. However, combining the two scrub purges may result in precipitation that could complicate the Cs/Sr removal system. Precipitation would result in a solids or sludge waste from which Cs and Sr could not be removed without redissolution. Also, the caustic scrub is expected to have lower cesium and strontium concentrations than the effluent from the acid scrub after ion exchange, hence there does not appear to be any benefit for combining the streams. Once melter and off-gas tests are able to provide better scrub solution composition data, tests of cesium and strontium removal of the scrub purge streams, over a range of acidity and with and without combining, should be done to optimize the scrub purge treatment system.

Testing of the cesium sorbent proposed in the baseline flowsheet has not been performed using INEEL waste simulants, and hence there is uncertainty regarding the performance of this sorbent. This uncertainty is the basis for alternative scheme 28, which would evaluate alternative neutralization agents to reduce the amount of sodium present in the ion exchange column feed, and scheme 29, which would replace the baseline sorbent with a different one.

The sorbent identified in the SBW Vittrification Feasibility Study^a for removing cesium from the acid scrub is IONSIV IE-95, and the baseline flowsheet retained this selection. The Feasibility Study sorbent selection was based on its successful use over a seven year period at the West Valley Demonstration Project (WVDP) to remove cesium from waste supernate.^b This sorbent or very similar sorbents have also been tested or used to remove cesium from waste streams at Savannah River, Oak Ridge, Three Mile Island, and Hanford.^c From 1963 to 1988, Savannah River generated a total of about 400,000 lb of spend Linde AW500 sorbent, which was used to remove cesium from evaporator overheads and other wastes, and is equivalent to UOP IONSIV IE-95.^d

At West Valley, approximately 550,000 gallons of nitric-acid based fuel reprocessing waste were neutralized with excess caustic. Neutralization resulted in a sludge layer forming and settling to the bottom of the storage tanks. Beginning in 1988, the liquid solution, called the supernate, was drawn off the tanks and processed by ion exchange using IONSIV IE-96. The pH of the supernate was about 10. (IONSIV IE-96 was produced in limited quantities for West Valley and Three Mile Island, and is presently commercially unavailable^e). IONSIV IE-95 is very similar to the former IE-96 product,

^a J. J. Quigley, B. D. Raivo, S. O. Bates, S. M. Berry, D. N. Nishioka, P. J. Bunnell, Feasibility Study for Vittrification of Sodium-Bearing Waste, INEEL/EXT-2000-00952, September, 2000; V. A. Descamp, C. L. McMahon, Vittrification Facility at the West Valley Demonstration Project, DOE/NE/44139-77, July, 1996.

^b J. J. Quigley, Trip Report of Visit to West Valley Demonstration Project, September 10, 1999.

^c L. A. Bray, K. J. Carson, R. J. Elovich, D. E. Eakin, HWVP Submerged Bed Scrubber Waste Treatment by Ion Exchange at High pH, PNNL-11033, March, 1996; S. M. DePaoli, D. T. Bostick, Process Wastewater Treatment with Hydrogen-Form CST and Chabazite Zeolite, ORNL/CP-98275; J. E. Miller, N. E. Brown, Development and Properties of Crystalline Silicotitanate (CST) Ion Exchanger for Radioactive Waste Applications, SAND97-0771, April, 1997; W. J. Dalton, Qualification Testing and Full Scale Demonstration of Titanium-Treated Zeolite for Sludge Wash Processing, DOE/NE/44139-72, June 30, 1997; J. A. Sundquist, J. C. Gillings, T. L. Sonntag, R. P. Denault, Bench-scale Treatability Testing of Biological, UV Oxidation, Distillation and Ion-Exchange Treatment of Trench Water from a Low-Level Radioactive Waste Disposal Area at West Valley, New York; D. C. Koopman, Factors Potentially Influencing the Tackiness of DWPF Streams, WSRC-TR-2000-00239, September 13, 2000; P. P. Murphy, J. S. Budkingham, Preliminary Laboratory Investigation for the Removal of Radioactive Cesium from Purex Aqueous Waste Solutions, RHO-CD-456, 1978.

^d C. M. Jantzen, Composition of Linde IE95 (AW500) Zeolite Fraction of Sludge, DPST-88-623, June 24, 1988.

^e Personal communication with Dennis Fennely of UOP, October 17, 2001.

containing more calcium and less sodium, both of which are minor components of both sorbents.^e) The spent sorbent was vitrified in the West Valley melter.

UOP IONSIV IE-95, and the very similar sorbents IE-96, Linde IE-95 and Linde AW500, are alkali metal aluminosilicates with partial substitution in the crystalline matrix by sodium, calcium and magnesium oxide. These sorbents are zeolites, a class of crystalline aluminosilicates with a highly regular structure of pores and chambers. Wastes processed with zeolite sorbents have been either alkaline or neutral, thus the baseline flowsheet includes neutralization of the acid scrub to a pH of prior to ion exchange. According to the manufacturer, the IE-95 can be used with wastes of pH 5 or higher.^f Bray^g references work he performed in 1990 which determined the Cs, Sr, and TRU distribution ratios for IE-95 and IE-96 as a function of sodium concentration, temperature and pH. This data, contained in an internal Hanford document, has been requested. After obtaining and reviewing the Hanford data on IE-95 performance, additional data needs relative to ion exchange sorbent selection can be determined.

As previously stated, the basis for the selection of IE-95 as the cesium sorbent in the baseline flowsheet was (1) its successful use in treating the West Valley supernate and (2) the successful vitrification of the spent sorbent in the WVDF melter. The greatest uncertainty relative to the sorbent is its performance on the neutralized acid scrub of the IWVF. The primary alternatives to zeolite sorbents are crystalline silicotitanate (CST) and ammonium molybdophosphate (AMP). Tests have been performed of cesium removal from SBW simulants using both CST and AMP.^h The acid scrub will differ in composition from SBW; however, the use of either of these sorbents would eliminate the need to neutralize the acid scrub prior to ion exchange. Vitrification of CST has also been demonstrated, although typically borosilicate glass has a TiO₂ limit of around 1.0%. The high phosphate content of AMP would also require either blending the spent sorbent with a large amount of SBW or calcine, reducing waste loading or using a special glass formulation.

Testing needs to be performed in order to validate selection of any ion exchange sorbent for the treatment of scrub solution in the IWVF process. Data is first needed from melter tests in order to define the range of scrub composition. Once the IX feed has been defined, testing of sorbents can proceed.

Caustic Quench and Scrub (Alternatives 30-31)

The caustic quench and scrub step cools the oxidizer effluent and removes acid gases, including HNO₃ vapor, HCl, HF, HI, and SO₂. Alternatives to the present single stage caustic quench/scrub step include a partial quench with water with no acid gas removal (Alternative 30) and quench and acid gas removal in more than one step (Alternative 31). The calcine vitrification feasibility studyⁱ assumes separate quench and scrub steps.

^f Personal communication with Dennis Fenely of UOP, July 13, 2001.

^g L. A. Bray, K. J. Carson, R. J. Elovich, D. E. Eakin, HWVP Submerged Bed Scrubber Waste Treatment by Ion Exchange at High pH, PNNL-11033, March, 1996.

^h N. R. Mann, T. A. Todd, K. N. Brewer, D. J. Wood, T. J. Tranter, P. A. Tullock, *Evaluation and Testing of IONSIV IE-911 for the Removal of Cesium-137 from INEEL Tank Waste and Dissolved Calcines*, INEEL/EXT-99-00332, April, 1999; N. R. Mann, T. A. Todd, *Removal of Cesium from Idaho Nuclear Technology Engineering Center Acidic Tank Waste Using IONSIV IE-911 Sorbent*, INEEL/EXT-2000-01570, September, 2000 (unpublished draft); T. J. Tranter, R. S. Herbst, T. A. Todd, H. B. Eldredge, *Evaluation and Testing of Ammonium Molybdophosphate-Polyacrylonitrile (AMP-PAN) as a Cesium Selective Sorbent for the Removal of Cs-137 from Idaho Nuclear Engineering and Technology Center Acidic Waste*, INEEL/EXT-99-00645, June 22, 1999.

ⁱ S. O. Bates, B. D. Raivo, J. J. Quigley, S. M. Berry, W. H. Landman, S. L. Palmer, T. M. Hipp, *Feasibility Study for Vitrification of Calcine in the Idaho Waste Vitrification Facility*, INEEL/EXT-01-00978, September, 2001.

The HCl concentration in the noxidizer off-gas is approximately one-tenth of the MACT limit for HCl/Cl₂ of 21 ppm, hence removal of HCl is not required. HF is expected to be at a comparable level or lower, and thus is not an emission concern. Based on present estimates, the SBW contains a total of about 0.2 Ci of ¹²⁹I. Based on information from Savannah River,^j all iodine in the melter vaporizes as NaI and then condenses as a submicron aerosol as the off-gas cools. Depending on DF factors for the condensed iodine aerosol in off-gas treatment steps, removal could be required to meet the ¹²⁹I Waste Acceptance Criteria (WAC) for the spent carbon waste. However, based on present assumptions, if no iodine is removed in the caustic scrub (by deleting the step) and all is captured on the GAC bed, the spent GAC will have an ¹²⁹I concentration of only about 5% of the Envirocare limit. Hence removal of iodine does not appear to be required.

Based on the present mass balance, the concentration of SO₂ in the noxidizer effluent is about 20 ppm, which equates to a total SO₂ emission of approximately 1 ton/year. According to EPA information for the year 1999, the Monsanto facility in Soda Springs emitted 7543 tons of SO₂, J. R. Simplot in Pocatello emitted 7123 tons, FMC Corporation in Pocatello emitted 2935 tons, the INEEL emitted 658 tons, Ash Grove Cement in Inkom emitted 489 tons, and the total SO₂ emissions for the state of Idaho was 23,671 tons.^k Assuming no change in INEEL emissions from other sources, the IWVF would add 0.15% to the SO₂ emissions if SO₂ is not removed. The actual SO₂ emission limit will be based on air quality modeling and how the emissions affect the air quality over Yellowstone and Grand Teton National Parks and Craters of the Moon National Monument. Recent modeling of emissions from the oil fired boilers at CPP-606 resulted in a requirement to limit the sulfur in the feed to 0.3%, lower than the statewide limit of 0.5%. However, based on the total possible rate of 1 ton/year SO₂, it's likely that no removal would be required.^l

Only trace amounts of nitric acid vapor are expected in the noxidizer effluent due to its destruction in the noxidizer. If not removed nitric acid emissions would be orders of magnitude lower than the emission limit of 0.333 lb/hr contained Idaho regulations, IDAPA 58.01.01.

The advantage of eliminating the caustic scrub step is the reduction in the amount of grout waste. The caustic scrub purge is about 50% as large as the neutralized acid scrub purge; eliminating the caustic scrub would reduce the amount of grouted waste by about 33%. However both acid and caustic scrub purge streams are small, 1-2 liter/hr; hence the reduction is not expected to have a significant impact on capital or operating cost. Although it appears the caustic tower could be eliminated, there is significant uncertainty in the partitioning of species such as chloride and sulfate that volatilize in the melter and are captured in scrub solutions. More scrub and off-gas analyses are needed in order to verify that the caustic quench and scrub tower can be eliminated.

If additional off-gas system test data or regulatory decisions indicate that removal of SO₂ or other acid gases is required, the question of single versus multiple stage removal becomes a mechanical issue with negligible impact to the mass balance and very little cost impact. Commercial caustic towers in industrial facilities typically have multiple sections, i.e., a packed bed scrub section on top of a quench section. Once requirements for removal are established, an optimal design can be developed. At that time other off-gas desulfurization technologies could be considered as well.

^j C. Randall, Iodine in HLW Off-gas, e-mail note to Bill Holtzscheiter, August 24, 2000.

^k Data available through EPA air data website <http://www.epa.gov/air/data/netemis.html>, Idaho statistics on webpage: http://oaspub.epa.gov/airsdata/net.ranking?geo=ID&cnty=+&pol=SO2&year=1999&rpp=25&fld=percent&fld=plnt_name&fld=addr&fld=state&fld=county&fld=year&fld=sic&fld=plantid&fld=lat_lon&fld=reg

^l Personal communication with John Gill, October 18, 2001.

GAC Bed Location (Alternative 32)

Because of the relatively large volume of spent carbon produced, 54 m³, it has been suggested to place the GAC bed downstream of the final HEPA filters to reduce the amount of radioactivity contained on the spent carbon waste. However the mass balances from the baseline flowsheet indicate that the radionuclide concentrations on the GAC bed are orders of magnitude less than limits of the most stringent disposal site, Envirocare. Unless changes in upstream off-gas treatment equipment result in higher levels of radioactivity on the carbon bed, there is no basis for changing the GAC bed and final HEPA filter arrangement.

NO_x Abatement Technology (Alternative 33)

Alternatives for NO_x abatement are discussed in the IWVF Off-gas Treatment Feasibility Study.^m The same report discusses the evaluation process that led to the selection of the noxidizer as the baseline NO_x abatement technology. In a value engineering session, the noxidizer flow scheme alternative scored higher than schemes that have the Studsviks process, SCR reactors, or SNCR reactors. However, in a separate evaluation of noxidizer technology, the SCR reactor was rated slightly higher than the noxidizer staged combustion technology.ⁿ Mass balance comparisons and capital cost estimates of the difference NO_x abatement schemes show advantages for the SCR reactor.

One unresolved issue that could affect the choice of NO_x abatement technology is the requirement for supplemental air to the noxidizer reduction chamber. The mass balance assumes this air rate is set by the reduction chamber temperature requirement. However, correspondence^o from John Zink, the noxidizer vendor, indicates that for burner stability, an air rate of 1900 scfm may be required. This air rate is 11 times greater than what is shown in the mass balance, and would significantly increase flowrates through the off-gas treatment equipment downstream of the noxidizer. Testing of an SCR reactor in the off-gas train of the Clemson Environmental Technologies Laboratory melter is planned and data from these tests should be reviewed to re-evaluate the selection of NO_x abatement technology.

Conclusions from Initial Screening

Based on the initial screening, the following alternatives can be eliminated from further consideration:

1. (3b) Total denitration
2. (20) Hg removal by ion exchange
3. (25) No treatment of acid scrub purge
4. (32) GAC bed downstream of final HEPA filters.

It is recommended that four changes be made to the baseline – blending SBW feed, use of granular sugar, removal of strontium from the neutralized acid scrub, and changing the melter plenum temperature to 500°C. Five schemes are evaluated in the next section based on results of the Visual Basic mass balance model. It is recommended that the evaluation of five other schemes be performed using the ASPEN mass balance model that will be developed this year:

1. (15) Cascaded acid scrub

^m R. Wood, D. Tyson, B. Bonnema, C. Olsen, A. P. Pinto, D. Wendt, S. Reese, B. Raivo, Feasibility Study for the Idaho Waste Vittrification Facilities Off-gas Treatment for Sodium-Bearing Waste, INEEL/EXT-01-00995, September, 2001.

ⁿ D. R. Tyson, "Vitrification Melter Off-Gas NO_x Abatement," EDF-IWVF-006, June 1, 2001.

^o L. Crynes, Revised Noxidizer Design, e-mail to S. J. Reese, January 17, 2001.

2. (18) Hg removal from scrub by electrolytic reduction
3. (19) Hg removal from scrub by sulfide precipitation
4. (21) Total scrub purge recycle
5. (22) No scrub purge recycle.

Test data from RSM-2 will also be used in some of these evaluations.

Two of the alternatives, a separate HEME wash water tank and multiple caustic quench/scrub steps, have little effect on the mass balance and can be deferred to later design studies. Additional test data is needed to evaluate the remaining 14 alternatives.

Evaluations Based on Mass Balance Results

Mass balances were determined for four SBW pretreatment schemes plus two variations of the baseline film cooler scheme. The results of these mass balances are discussed in the following sections. Of all the process alternatives, pretreatment schemes have the greatest potential for significant savings because they affect all downstream flowrates.

Barium Sulfate Precipitation

A mass balance was determined for the barium precipitation scheme assuming an 80% removal of sulfate and 20% excess barium, based on data in Reference 17. SBW and barium nitrate would be fed to a precipitation tank and mixed for approximately one-hour. The mixture would then be filtered, with the filtrate sent to melter feed mix tanks, and the filtered solids periodically removed from the filter and packaged as secondary waste. A solids removal efficiency of 99% was assumed for the filter.

The mass balance assumes a glass waste loading of 35%. Based on the results of FY-01 glass melt tests in which waste loading was varied between 25% and 45%, Peeler et al^p concludes that if not constrained by a sulfate salt layer, the waste loading in SBW glass would be constrained by PCT release or nepheline formation upon cooling at a waste loading greater than about 40%. Thus a waste loading higher than 35% may be achievable if sulfate is removed from the feed. However, 35% was assumed to be conservative.

Assuming a 2 molar barium nitrate solution is used to precipitate sulfate, and that the processing schedule is unchanged, the total melter feed rate would increase by 2% and the glass rate decrease by 36% from rates of the baseline mass balance. The sulfur concentration in the glass would decrease by nearly 70% relative to the baseline mass balance. Off-gas and liquid scrub rates would remain unchanged.

The savings in disposal costs from the reduced glass volume are approximately \$200 million, based on disposal at Yucca mountain.^q Additional savings would result from a smaller melter or reduced processing schedule. The savings would be partially offset by equipment to precipitate barium sulfate, and treat and dispose of the resultant solids. Approximately 45,000 kg of solids would be produced, including barium sulfate precipitates, small amounts of other precipitates and SBW UDS separated by the filter. If packaged in a dry form, about 30 m³ of TRU waste would be produced. If grouted at a 25

^p D. K. Peeler, T. B. Edwards, I. A. Reamer, R. J. Workman, J. D. Vienna, J. V. Crum, M. J. Schweiger, Glass Formulation Development for INEEL Sodium-Bearing Waste (FY-2001 WM-180), WSRC-TR-2001-00295, September 21, 2001

^q Disposal cost of \$540 K per 10-ft canister (0.62 m³ glass) were taken from Appendix F of the Idaho High-Level Waste & Facilities Disposition Draft Environmental Impact Statement, DOE/EIS-0287D.

wt % solids waste loading, about 100 m³ of waste would be produced. Using WIPP disposal costs consistent with the Idaho High-Level Waste and Facilities Disposition Draft EIS,^r these two waste volumes equate to disposal costs of about \$8 million and \$25 million respectively. An alternative treatment and disposal path for these solids is to perform the precipitation in the Tank Farm tanks, letting the precipitates settle to the bottom of the tanks and be disposed of with the heel solids.

Siemer (Reference 18) determined that the precipitate would contain 1-2% of the plutonium in the SBW. Assuming the precipitate would contain 1% of all TRU elements in the SBW liquid, the precipitate would have a TRU concentration of 20,000 nCi/g. Additional TRU will be contained in the solids from UDS in SBW; the total TRU concentration in the waste solids is estimated to be 40,000 nCi/g for the dry waste form or 5,000 nCi/g for the grouted waste. The solids chemical composition is given in Table 5.

Table 5. Barium precipitation solids composition.

	wt %		wt %
Ag+1	0.001	NO3-1	8.56
Al+3	1.15	Pb+2	0.01
As+5	0.0001	Pd+3	0.01
B+3	0.01	PO4-3	3.28
Ba+2	48.7	Ru+4	0.007
Ca+2	0.08	Se+4	0.03
Cd+2	0.003	Si+4	0.41
Cl-1	0.02	Sr+2	0.001
Cr+3	0.01	SO4-2	34.36
Cs+1	0.01	V+5	0.0002
Cu+2	0.003	Zn+2	0.004
F-1	0.001	Zr+4	0.55
Fe+3	0.40	Be+2	0.00004
Gd+3	0.002	Ce+4	0.001
Hg+2	0.18	Co+2	0.0003
K+1	0.29	Nb+5	0.20
Li+1	0.003	Sb+4	0.001
Mg+2	0.03	Sn+4	0.04
Mn+4	0.03	Ti+4	0.02
Mo+6	0.007	Tl+3	0.03
Na+1	1.54	U+4	<u>0.02</u>
Ni+2	0.005	Total	100.000

Based on the magnitude of the potential savings for the barium precipitation scheme, testing is recommended to confirm the glass waste loading for a reduced-sulfur SBW surrogate, and to develop a waste form for the barium sulfate precipitate.

SBW Absorption on Silica Gel

In this scheme, SBW would be absorbed on silica gel and the resultant slurry would be fed to an evaporator in which it would be evaporated to dryness. Kirkham provides data from tests of this scheme in Reference 2. These tests showed that the solids produced are flowable at waste loadings up to 84%, and at high waste loadings and a temperature of 140°C, evaporation resulted in a loss of 76-78% of the mass of the original SBW.

^r \$205 K per 0.8 m³ cask.

A comparison of the simulant composition used by Kirkham to the total SBW composition is shown in Table 3.

Table 3. Comparison of test surrogate and projected SBW composition.

Species	Test Surrogate Concentration, Mol/liter	Projected Total SBW Concentration, Mol/liter	Ratio, Surrogate/Present Projected Composition
H ⁺	1.94	2.4	1.24
Al ⁺³	0.784	0.557	0.71
Ca ⁺²	0.576	0.514	0.89
K ⁺	0.245	0.164	0.67
Na ⁺	2.36	1.52	0.64
NO ₃ ⁻	6.95	5.93	0.85
SO ₄ ⁻²	0.0637	0.0506	0.79

Because Kirkham's simulant was more concentrated than the present projected SBW composition, a higher fractional loss on evaporation could possibly be expected with SBW, as currently projected.

The present baseline mass balance for the total SBW case shows consumption of 813,000 kg of SiO₂, the primary glass forming component. The baseline mass balance assumes the GFCs are 65% SiO₂; a more recent glass formulation increases the SiO₂ content to 68%.^s A mass balance was calculated for the silica gel absorption scheme assuming consumption of the same quantity of SiO₂, 813,000 kg, as the present baseline mass balance. This is equivalent to a SBW waste loading on silica gel of 85%. The mass balance also assumes 80% of the original SBW mass lost in evaporation, well within the range seen in Kirkham's tests considering the feed composition differences. The evaporation mass loss is equivalent to the loss of all the water and initial nitric acid in the SBW plus the loss of an additional 23% of the nitrate through reactions of silica gel with the waste that produce additional nitric acid.

A glass waste loading for the SBW absorption scheme was unchanged from the baseline flowsheet, hence the glass rate is unchanged. With less nitric acid and nitrate in the dry SBW feed, the rate of sugar was reduced by 80% for the SBW absorption scheme mass balance. Also, granular sugar was used rather than a sugar solution. Other mass balance comparisons are shown in Table 4.

Table 4. SBW absorption scheme flowrate comparison.

	SBW absorption Scheme	Baseline Scheme	Ratio SBW absorption/ Baseline
Melter off-gas rate, sm ³ /hr	68	814	0.083
Film cooler effluent rate, sm ³ /hr	166	1,915	0.086
Quench column feed rate, sm ³ /hr	311	2,060	0.15
Quench liquid rate, liters/hr	3,185	20,477	0.16
First HEPA bank effluent, sm ³ /hr	316	2,057	0.15
Noxidizer effluent rate, sm ³ /hr	809	3,356	0.24
Caustic scrub rate, liters/hr	4,044	16,780	0.24
Off-gas to stack, sm ³ /hr	809	3,359	0.24

^s J. D. Vienna, SBW-22, e-mail to Keith Perry, July 12, 2001.

Based on the flowrate ratios in Table 4, savings of 50-70% could be expected in off-gas system equipment.^t The cost estimate for the baseline off-gas treatment is \$18.2 million, as per the Off-gas Feasibility Study.^u Thus the potential savings from reduced off-gas equipment costs are \$9-13 million. Additional savings should be realized from a significant reduction in facility size due to reduced off-gas treatment equipment space requirements.

The SBW absorption scheme would have additional costs over the baseline due to (1) feed treatment (evaporation) (2) incremental costs of silica gel relative to silica, and (3) costs to treat and dispose of the recovered nitric acid. Incremental feed treatment costs are expected to be small. Wendt and Haefner (Reference 1) estimated cost savings in feed treatment for SBW evaporation to dryness compared to the baseline scheme. Based on typical costs for bulk silica gel, the total cost for the silica gel would be about \$500 K.

About 880,000 gallons of evaporator condensate, with a nitric acid concentration of 3.5 molar, would be produced from the evaporation of the absorbed SBW. Concentration of the condensate in the LET&D would reduce the volume to about 310,000 gallons of 10 molar acid. If neutralized with caustic and packaged as dried sodium nitrate waste, about 3,300 55-gal drums would be produced. Disposal of this waste at Hanford would cost about \$600 K.^v Energy costs for evaporation of SBW plus evaporation of the condensate in the PEW followed by concentration in the LET&D, are estimated to be about \$100 K, based on an energy cost of \$5/million Btu.

Based on this initial review, the silica gel absorption process appears to offer significant savings in facility costs by reducing the size of the off-gas treatment system. Further development is recommended. Data is needed on impurities in the SBW evaporator condensate in order to better determine requirements for treating the condensate and disposing of the final waste. Melt tests are needed to confirm that the silica gel has no detrimental effect on glass properties and to optimize waste loading in the glass. At the present time, the glass waste loading appears to be limited to approximately 20% due to the sulfur content of the waste; a higher waste loading would result in significant savings due to reduced glass waste disposal costs (See section on barium sulfate precipitation). However, Kirkham² prepared three glasses from his evaporated SBW-silica gel products, using higher waste loadings (28-40% based on SBW oxides). Should higher waste loadings be feasible because of sulfate removal or through increased sulfate retention with the absorbed SBW, additional evaporation tests would be needed to confirm flowability of the dried melter feed. Melter tests are also needed to better determine the effects of the dry feed on volatile and entrained particulate concentrations in the melter off-gas. Finally, scale tests of SBW evaporation equipment would be needed prior to final design.

Partial SBW Evaporation

To evaluate partial SBW Evaporation, mass balances were prepared for two different schemes. From SBW evaporation test results of John McCray (References 4-5) it appears that 70-80% of the SBW volume could be evaporated without forming a product that is sticky or sets up upon cooling. Assuming 73% volume reduction, an ASPEN Plus simulation indicated that about 82% of the water and 80% of the nitric acid in the evaporator feed would be removed. The resultant concentrate would be a slurry

^t Using the 0.6 power rule with reduced capacity factors, the reduction would be 57%-78%. However, considering the low flowrates for the new scheme and the uncertainties in the mass balance, the range was rounded down to 50-70%.

^u R. Wood, D. Tyson, B. Bonnema, C. Olsen, A. P. Pinto, D. Wendt, S. Reese, B. Raivo, *Feasibility Study for the Idaho Waste Vittrification Facilities Off-gas Treatment for Sodium-Bearing Waste*, Appendix K, INEEL/EXT-01-00995, September, 2001.

^v Based on \$1,900 per 40,000 lb truckload plus \$710/m³.

containing precipitated nitrates of sodium and aluminum. For one mass balance, this slurry was then assumed to be fed to the melter. A second mass balance was based on filtration of and separate treatment of the solids. For this second case, the melter feed is thus only the liquid concentrate.

The mass balance for the slurry feed scheme shows 50-60% reduction in off-gas flowrates upstream of the noxidizer and 25-30% downstream of the noxidizer compared to the baseline mass balance. The reduced flowrates would result in capital cost savings from smaller off-gas equipment and reduced plot space requirements. From evaporation of SBW, 2.7 million liters of condensate would be produced. If processed through the LET&D, 180,000 gal of 10 molar nitric acid would be produced. The acid would contain small amounts of fluorides, mercury and chlorides. If neutralized with caustic, dried and grouted, about 2,000 drums of waste would be produced. The savings, as well as the incremental costs, from this partial evaporation scheme are not as great as for the total evaporation (silica gel absorption) scheme discussed above. However, the scheme offers a means to significantly reduce off-gas flowrates without changing to a solid melter feed system.

John McCray's SBW evaporation/precipitation test results (Reference 5) indicate that as SBW is sequentially evaporated and cooled, first sodium nitrate crystallizes and then aluminum nitrate. Based on his results, perhaps 70% of the sodium nitrate could be crystallized before significant amounts of other species precipitated. Reducing the level of sodium alone or both sodium and aluminum would not significantly reduce the volume of glass produced because the glass waste loading, and hence the glass volume, is controlled by sulfate in the waste. However, if sulfate were also removed from the waste, reducing the amount of sodium and aluminum would be beneficial.^w

A sketch of this scheme is shown in Figure 1.

The mass balance for the sulfate/nitrate precipitation scheme is based on the following assumptions:

1. 70% sulfate removal by barium precipitation
2. 1.2 moles barium nitrate added per mole of sulfate in SBW
3. 100% removal of BaSO_4 precipitate along with UDS in SBW by filtration
4. 73% volume reduction by evaporation (81.5 wt % removal of H_2O , 80 wt % removal of HNO_3)
5. Precipitation upon cooling of 80% of the NaNO_3 and 55% of the $\text{Al}(\text{NO}_3)_3$ in the waste
6. 100% removal of the precipitated nitrates
7. 35% waste loading in the glass.

The mass balance results show, relative to the baseline mass balance:

1. 91% reduction in the waste feed rate
2. 75% reduction in the melter feed rate
3. 70% reduction in the glass rate
4. a slight reduction (2%) in the sulfate concentration in the glass
5. 70% reduction in the melter off-gas rate
6. 65% reduction in quench, scrub and noxidizer feed off-gas rate
7. 65% reduction in quench and scrub liquid rates
8. 40-45% reduction in off-gas flowrates downstream of noxidizer
9. Generation of about 30 m^3 (150 55-gal drums) of TRU waste
10. Generation of about 660 m^3 (3,300 55-gal drums) of Hanford Category 3 waste.

^w Personal communication with John Vienna, October 25, 2001.

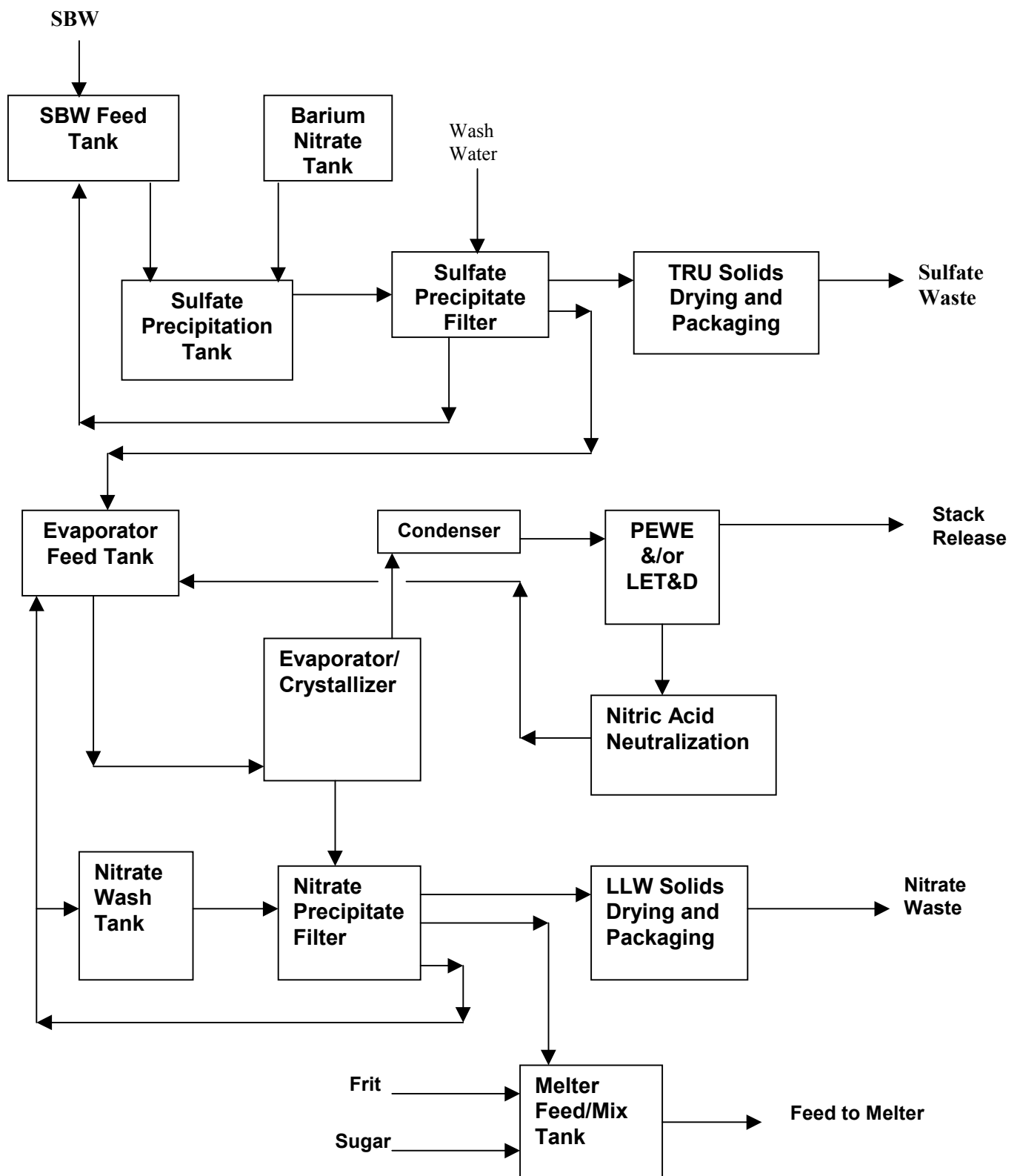


Figure 1. Sulfate/Nitrate Precipitation Pretreatment Scheme.

The estimated savings in HLW glass disposal costs is about \$380 million. The estimated cost of disposal of the TRU waste at WIPP is \$8 million, and of the LLW at Hanford is \$5 million. The waste volumes are based on dried precipitates; if the precipitates were grouted instead of dried, the volumes (and disposal costs) would increase. The cost estimate for disposal of LLW assumes charges for Category 3 waste. Data in Reference 5 shows 0.1-0.25 ml of entrained liquid per g of precipitate. Assuming a value of 0.2 ml entrained liquid per g of solids and 95% removal of the entrained liquid per wash, a single wash will bring the TRU concentration in the solids to about 30 nCi/g. The waste would be Category 3 due to concentrations of ^{137}Cs , ^{90}Sr , ^{238}Pu , ^{239}Pu , and ^{241}Am . A second wash of the same efficiency would bring the actinide concentrations to less than 2 nCi/g and below Hanford Category 1 levels, but ^{137}Cs and ^{90}Sr would still be above Category 3 limits. A wash removal efficiency greater than about 97% would be required to remove ^{137}Cs to below the Category 1 limit and about 99% to remove ^{90}Sr to below Category 1 limits. Additional tests are needed to better determine the classification of this waste.

Additional savings for the sulfate/nitrate scheme would result from the reduced off-gas flowrates.

As previously stated, the assumed SBW waste loading in glass has dropped from 36% to 20% in the past two years. This change increases the glass waste volume by 80%, and increases the disposal cost by \$200 million. Sulfate precipitation offers a method to reduce the glass volume back to that assumed in the FY-2000 Feasibility Study.^x Sulfate precipitation combined with evaporation has the potential to not only reduce the cost of the IWVF back to near the Feasibility Study estimate, but to reduce it by another \$200 million or so.

Chilled Air Film Cooler and Transpiring Wall Cooler

The baseline mass balance assumes inlet air to the film cooler is 60°F and the film cooler air to steam ratio is 1.4. If the air to steam ratio is increased to 10 and the air is cooled to 41°F, off-gas rates downstream of the film cooler are reduced by 7-8%. More detailed design calculations and cost estimates would be required to determine if the savings from the reduced off-gas flow compensate for the cost of chilling the film cooler air. However, since the savings appear to be low, chilling the film cooler air is not recommended at this time.

Replacing the film cooler with a transpiring wall cooler results in a 54% reduction in off-gas flow upstream of the noxidizer and about a 40% reduction downstream of the noxidizer. More feed air is required in the noxidizer reduction section for the transpiring wall cooler case because less oxygen is contained in the noxidizer feed. A comparison of noxidizer feed and effluent streams is shown in Table 6.

^x S. O. Bates, B. D. Raivo, J. J. Quigley, S. M. Berry, W. H. Landman, S. L. Palmer, T. M. Hipp, Feasibility Study for Vitrification of Calcine in the Idaho Waste Vitrification Facility, INEEL/EXT-01-00978, September, 2001.

Table 6. Effect of replacing film cooler with transpiring wall cooler.

	Noxidizer Feed		Noxidizer Effluent	
	Baseline	Trans. Wall	Baseline	Trans. Wall
sm ³ /hr	2,057	1,104	3,356	2,076
	Mole Fraction	Mole Fraction	Mole Fraction	Mole Fraction
CO ₂	0.029	0.054	0.071	0.081
CO	0.004	0.007	0.000	0.000
H ₂ O	0.496	0.643	0.559	0.562
N ₂	0.360	0.217	0.356	0.344
NO	0.014	0.026	0.001	0.001
NO ₂	0.010	0.019	0.000	0.000
O ₂	0.081	0.028	0.009	0.009
HNO ₃	0.001	0.002	0.000	0.000
Ar	0.004	0.003	0.004	0.004

Rates and compositions of secondary waste streams are expected to be essentially the same for the transpiring wall cooler compared to the baseline flow scheme. Given no pretreatment scheme that reduces melter off-gas rates, the magnitude of the reduction in off-gas rates would justify further study to better evaluate the evaporative cooler design, i.e, Studsvik-type, SCWO-type or other design, and then to test the preferred design in order to demonstrate performance. However, if an evaporation pretreatment scheme is adopted as recommended above, no change is recommended at this time to the baseline film cooler design. Off-gas rates for evaporation schemes are in range of 60-240 sm³/hr; air or steam requirements to cool this off-gas in a film cooler are similarly small, and hence saving from a different type of cooler would be small.

cmb

cc: C. M. Barnes Letter File CMB-11-01
L. Lauerman, MS 5218
W. H. Landman, MS 3211
A. L. Olson., MS 5218
D. D. Taylor, MS 3779

Uniform File Code: 6153.102

Disposition Authority: ENV1-k2b

Retention Schedule: EPI* Cutoff after project/program completion, cancellation or termination or in 5 year blocks. Retire to FRC 2 years after termination of project/program. Destroy 25 years after termination of project/program.

NOTE: Original disposition authority, retention schedule, and Uniform Filing Code applied by the sender may not be appropriate for all recipients. Make adjustments as needed.